DEEP SEA DRILLING PROJECT TECHNICAL REPORT No. 4

DRILL STRING Drill Pipe Bumper Subs Drill Collars Coring Equipment

CONTRACT NSF C-482

PRIME CONTRACTOR THE REGENTS, UNIVERSITY OF CALIFORNIA

SCRIPPS INSTITUTION OF OCEANOGRAPHY University of California at San Diego

DRILL STRING

Drill Pipe Bumper Subs Drill Collars Coring Equipment

Prepared For

National Science Foundation Under Provisions of Contract NSF C-482

By

Deep Sea Drilling Project Scripps Institution of Oceanography University of California at San Diego

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INTRODUCTION

Many new and complicated technical systems had to coalesce on D/V Glomar Challenger to enable the Deep Sea Drilling Project to enjoy the continuous success it has recorded todate.

Dynamic positioning worked beyond fondest expectations and the problem of re-entry was overcome with little difficulty, but without a good, tough drill string - some of which is still in service after more than three years of drilling and coring in a hostile environment there would be no Deep Sea Drilling Project.

Included in this Technical Report are extensive discussions on drill pipe, bumper subs, drill collars and coring equipment.



Glomar Challenger drilling crew rigs hydraulically operated power sub, center, preparatory to commencing drilling operations at a deep water drilling site in the Gulf of Mexico during the first leg of the Deep Sea Drilling Project. At right is the drawworks and, on opposite side of derrick floor is the specially designed core line reel used in the retrieving of the core barrel. Beyond derrick floor is Global Marine designed automatic drill pipe racker which holds 24,000 feet of 5-inch drill pipe. View, looking forward, was taken from wheel house of vessel.

While on station, the drill string is suspended from D/V Glomar Challenger with no lateral support between the keel and the ocean floor. Drilling and coring are accomplished by turning the string at the top with a power sub. The heave, roll and pitch of the drilling vessel add additional stresses.

Since drilling and coring are done without using a riser system for drilling mud returns, raw sea water is normally used, further increasing the stress on the pipe.

Corrosion inhibitors are used, but the automatic pipe racks are above deck, exposing the pipe to a moist marine environment while in storage.

Sites have been cored with surface observed currents of four knots. Subsurface currents are not measured. Even with these hazards, drill string losses todate have been caused primarily by operational problems.

Global Marine Inc. designed the drill string to be part of their answer to the Project's Request for Proposal. This design, with minor modifications, was the one taken to sea. As operational problems arose, the Deep Sea Drilling Project and Global Marine Inc. mutually agreed on modifications.

The initial Program Plan attempted to schedule D/V Glomar Challenger to drilling sites during the most favorable weather conditions and to stay within the 30 degree parallels. As drilling crews gained confidence, sites were successfully worked in the North Atlantic and the North Pacific. A further three-year extension would take the Challenger into the Arctic and Antarctic Oceans.

With this increased confidence in the drill string and an upgraded program for the dynamic positioning system, it would be possible to drill in water depths less than 3,000 feet. This capability would open vast new areas of scientific interest.

Scripps Institution of Oceanography, of the University of California at San Diego, is managing institution for the Deep Sea Drilling Project. The Project is funded by the National Science Foundation, an agency of the United States Federal Government and is a part of the Foundation's National Ocean Sediment Coring Program.

Actual drilling is being done by Global Marine Inc., of Los Angeles, California, under contract to Scripps Institution of Oceanography.

Advice regarding scientific planning for the Deep Sea Drilling Project is being provided by panels whose members are broadly representative of the nation's scientific community drawn from many Universities, Government Agencies, and industrial organizations.

The Joint Oceanographic Institutions for Deep Earth Sampling (JOIDES), consisting of Lamont-Doherty Geological Observatory of Columbia University, Woods Hole Oceanographic Institution, The Rosenstiel School of Marine and Atmospheric Sciences, of the University of Miami (Florida), The University of Washington (Seattle), and Scripps Institution of Oceanography, assisted in the formulation of plans for the Project. JOIDES panels still give scientific advice and guidance to the Deep Sea Drilling Project.

Many individuals from the petroleum industry have also furnished technical advice and assistance to the Project.

April, 1972

ACKNOWLEDGEMENTS

The Deep Sea Drilling Project gratefully acknowledges the contributions of many companies and individuals to the successful adaptation to drilling in the deep oceans of the various drill string components – drill pipe, bumper subs, drill collars and coring equipment – which have so far been responsible for recovering more than 10 miles of priceless sedimentary core material from beneath the deep ocean floor after 21 expeditions of the Project.

Our thanks go to:

Mr. Jack Reed and Dr. Thad Vreeland, of Global Marine Inc., for design and analysis of the drill string configuration;

Mr. Alex Maradudin (deceased), formerly with Standard Oil Company of California, for review and supervision of manufacture of the drill pipe;

Dr. Arthur Lubinski, of Pan-American Petroleum Corporation's Research Division, for review of the proposed drill string configuration;

The Petroleum Industry Technical Assistance Panel for its valuable advice;

DRILCO for review of proposed drill collar tool joint change;

Hycalog, Inc. for the design and expediting of basic coring equipment and the training of Project personnel;

Christian Diamond Products Company for design of alternate coring methods and counsel for appropriate Project personnel;

Project Engineer Darrell L. Sims for engineering management.

We acknowledge the continuing help of Mr. A. R. McLerran, National Science Foundation Field Project Officer with the Deep Sea Drilling Project and the support of the National Science Foundation.

The support and guidance of the Joint Oceanographic Institutions for Deep Earth Sampling (JOIDES) in fostering the Deep Sea Drilling Project is fully acknowledged.

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M.N.A. Peterson Co-Principal Investigator Deep Sea Drilling Project

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DRILL PIPE

DRILL PIPE

The statement of work in the Request For Proposal for Deep Sea Drilling Project drill pipe was quite short - "Two 23,000-foot drill strings, one working and one spare should be aboard, with all necessary collars, bits, core barrels and accessory items." Steel, non-ferrous or combination strings were considered. A 2 1/2-inch core encased in a plastic liner was requested. The proposal was to give the type, grade, diameter, weight per foot and length of stands, and to include an analysis of this design based on relevant operating conditions.

Global Marine Inc., selected to be the subcontractor, proposed 5-inch outside diameter, 19.6 lb. per foot grad S-135 external upset steel pipe fitted with modified 5 1/2-inch full hole flash-weld, 18-degree taper tool joints. A detailed analysis of this string conducted by Dr. Thad Vreeland, of Pasadena, California, for Global Marine Inc. concluded this string would give excellent life up to and including the design depth of 22,500 feet total drill string length.

The Deep Sea Drilling Project enlisted the assistance of major oil companies to review the proposal. The then Technical Assistance Panel, composed of representatives from major oil companies, agreed that the proposal was acceptable. Dr. Arthur Lubinski and his group at Pan-American Petroleum Corporation Research Division in Tulsa, Oklahoma, reviewed Dr. Vreeland's mathematical model and advised they were in agreement.

Global Marine Inc. was so advised and placed their order. The pipe was rolled by Jones and Laughlan Steel Company, of Pittsburgh, Pennsylvania, and the tool joints made and installed by Hughes Tool Company, of Houston, Texas.

The services of Mr. Alex Maradudin (now deceased) were obtained by the Project to review the Purchase Order for mill and test procedures. Mr. Maradudin, a retired metallurgist from Standard Oil Company of California's Research Laboratory, La Habra, California, and an expert in oil field tubular goods, advised that the Purchase Order was technically acceptable.

To insure the minimum of manufacturing defects, Mr. Maradudin was retained by Global Marine Inc. to supervise the manufacture and inspection of the S-135 drill pipe.

The drill pipe was cleaned and sandblasted after the tool joints were attached and given final sonoscope and end area inspection. The outside was coated with a zinc-rich organic coating and the inside coated with corrosion preventative oil.

The drill pipe was not plactic-coated on the inside. The Project and Technical Assistance Panel felt the coating could become brittle at the low temperatures of the deep ocean and that extensive coring would scar and peel the coating.

The external coating definitely has retarded corrosion. External corrosion was first noticed on Leg 12, or two years after the pipe was placed in service.



External corrosion of 5-inch drill pipe at least two years after being placed in service.

On the first few runs in deep water, the external coating seemed to form blisters. When the pipe was pulled up in shallower water, these blisters would flake off, making enough acoustical noise to confuse the positioning system.



Blisters in external coating apparently caused by water at depth being trapped.

Hydraulically installed rubber drill pipe protectors were installed at 5-foot intervals along the single 30-foot stands of pipe. This was done to reduce resonant motion, to protect the drill string during automatic pipe-racking operation, and to provide color coding of the protector rings as a reference system for identification of individual lengths. Although this color code would have made it possible to establish stress exposure of the pipe, it was operationally impossible to establish and follow such a program. As the Deep Sea Drilling Project progressed, stress fatigue did not appear as a problem, so interest in such a program diminished.





Color coded drill pipe protectors at 5 foot intervals on the drill string.

Mechanical pipe racker used to store the drill string.

Thread protectors were installed and the pipe transported to Orange, Texas, where it was loaded aboard D/V Glomar Challenger after builder's trials. When placed in the automatic pipe racker, each joint of the pipe was made up and broken down several times with a high zinc-content thread lube to prevent galling.

The drill pipe is run into and pulled out of the hole with an IDECO Dual Elevator system. One elevator is used to pull the pipe in the conventional manner. The second set replaces the drill pipe slips and supports the pipe on the rotary table. As the drill pipe is laid down in the pipe racker, the No. 1 elevator has been lowered back to the drill floor. This elevator is set to one side and the links that attach the elevators to the traveling block are removed and hooked onto the No. 2 elevator. This elevator now becomes the No. 1 elevator and the string is lifted and set off on the remaining elevator, thus the elevators are leap-frogged while pulling or running the pipe.

This system eliminates marking and necking the pipe with slips. As S-135 is notch sensitive, eliminating slip marks undoubtedly has extended the drill pipe life.

The introduction of re-entry to the Program Plan also introduced several problems involving the drill pipe. The actual use factor of re-entry was an unknown, but if re-entry was to be used frequently, it would be operationally desirable to run the cones and bases through the moon pool. This would require either removing or modifying the drill pipe supporting expotential horn. As re-entry was specifically eliminated from the original Request For Proposal, and casing was the only other tubular goods to be run through the horn, the opening at the top of the fixed horn was 30 inches. In addition, the Project had received requests to drill sites that required a drill string longer than 22,500 feet.

A subcontract was entered into with the Southwest Research Institute to determine the feasibility of these questions. Their report indicated drill pipe support, such as given by the expotential horn, is necessary. The present horn is structurally over designed and could be a lighter removable structure. The "piccalo" or scabbard-type support could be shortened from 80 to 50 feet and given necessary support to the drill pipe. The report, in addition to supporting Dr. Vreeland's work, extended computed estimates of drill pipe life for a 30,000 foot string. The computer program for this report is available if modification or review of the drill pipe is required.

The Request For Proposal included pipe inspection at frequent intervals as a requirement, but did not specify the type of inspection. Global Marine Inc. officials stated in their reply they would furnish magnetic particle and Magna Glo ultra violet inspection for tool joints and heavy sections. A sonic (assumed to be Sonoscope, trade name of Tuboscope's electromagnetic inspection equipment) pipe inspection unit would be used to inspect the pipe in the pipe racker. The subcontract retained the magnetic particle (Magna Glo or similar) inspection for tool joints, bumper subs, drill collars and heavy sections and specified inspecting the drill pipe with a magnetic flux-type instrument similar to Tuboscope equipment.

Global Marine Inc. had planned to use Tuboscope's well-head isolog which would allow inspection on each trip. This unit is placed on the rotary table and the pipe inspected as the string is pulled out of the hole. This is an external unit with the coils and detector placed around the pipe. The dual elevator system would have required placing the unit under the rotary table, however, Dr. Vreeland's conclusion that support was needed through the expotential horn of the drill pipe with rubber protectors made inspection with an external system very undesirable, as the coils and detector head would have to be opened and moved across each protector. In addition, this well-head unit was in the developmental stage and results still questionable, as the pipe had to be centered as it is pulled through the unit. Inspection in the pipe racker with the external sonoscope would have been so slow as to be prohibitive.

An industry search for an internal sonoscope indicated Tuboscope was the only company interested. Tuboscope had designed and tested for Phase II of Mohole, a prototype internal sonoscope. The sonde was to be run on the 7-conductor logging line so the string could be inspected while in the hole. Tuboscope advised they could furnish a more sensitive and dependable sonde if inspection was limited to one stand at a time (100 feet). This would allow inspecting the pipe in the racker even though the racker supported the pipe at three places with steel fingers. This blip would be put on the tape and easily identified.

Tuboscope built the system and the shipboard equipment was installed at Hoboken, New Jersey, at the end of Leg 1 in 1968. The pipe was inspected then and has been inspected at the end of all succeeding legs. Todate (end of Leg 20) inspection seems to be adequate. There have been no drill pipe failures due to flaws or stress corrosion. On Leg 18, a crack was visually detected in a joint of pipe approximately three feet above the pin. The crack was 7/8-inch long and apparently was caused by fatigue. It is not known if this joint of pipe had been in service from the start of operations in 1968. In fact, the only drill pipe failure was caused by trying to core without bumer subs in the bottom hole assembly.

Suspect joints are removed from the string on the next site as the string is lowered. At first, these were inspected at the next port by optical boroscope. Accumulated scale in the pipe made this a slow process, so an ultrasonic instrument was placed on board and now the drilling crew checks the suspect joints.

The inspection after Leg 11 indicated several joints had internal pitting. Four joints with the deepest indentations were carefully inspected and the points marked. The joints were shipped to Los Angeles, California, where they were inspected, using the external sonoscope. A comparison of inspection records indicated the external sonde was located pits that should be checked with either the boroscope or supersonic probe. Depths of pits were measured with the ultrasonic probe. One joint was then cut into coupons and sent to several major oil company laboratories for examination. The concensus of opinion was the pitting was caused by aerated salt water.



Drill pipe coupon used to check sonoscope inspection accuracy. Several procedures are used to control corrosion. The pipe is flushed with fresh water just before pulling and a slug of mud is pumped in and a slug of cationic oil soluble corrosion inhibitor is added on top of the mud. After racking, a pipe rattler is used to loosen the scale so the inhibitors can wet the pipe. Tight thread protectors are used on the pin end of the stand to reduce sweating and keep sea water out during rough weather. The pin end faces forward. Corrosion measuring rings are run at several spots in the string to watch for any change in the rate of metal loss.

When the scientific community became aware that one of the legs on the first extension granted the Deep Sea Drilling Project would be in the Mediterranean Sea, several scientists expressed interest in a site located in the Black Sea. The Project was asked to estimate logistics.

As the Black Sea has an appreciable H_2S content, the drill string would be one area that could be affected.

An industry and literature survey indicated S-135 drill pipe could be subjected to hydrogen embrittlement if the site in the Black Sea were drilled. Using Grade "E" drill pipe would eliminate this problem, but the cost would be prohibitive to transport sufficient pipe and change it out twice on D/V Glomar Challenger. Two rules of thumb established by the oil industry to select steel tubular goods for H₂S environment are:

- Elongation should be 23% to 25% after quench and temper since rolled goods usually have local hard spots.
- (2) Keep the yield point of steel below 80,000 psi and the hardness below Rc 22.

In addition, the maximum length of drill string required in the Black Sea would be 10,000 feet, or half the designed length of the Deep Sea Drilling Project drill string. The H₂S concentration increases with water depth, so it would be affecting the section of drill string with lower stress levels.

A review of these factors indicate the Deep Sea Drilling Project's S-135 drill pipe could be used to drill in the Black Sea. It should be noted that the hydrogren in the steel is released if exposed to the atmosphere, so it is not permanently changed.

Despite these studies that indicated the feasibility of using S-135 drill pipe in hydrogensulfide laden situations, drilling was not attempted in the Black Sea during Phase II because scheduling of D/V Glomar Challenger to undertake such drilling would have permitted an unsatisfactorily small amount of time for the Mediterranean Program.

In summary, the first three years of operations have proved the drill pipe to be more than adequate for the water depths and sea states operated in . Also, using a continuous drill pipe string has simplified handling, racking, inspection and logistics.

Although pitting is evident and will increase with use, it is not possible to project the useful life of the drill pipe.

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BUMPER SUBS

BUMPER SUBS

In the Request For Proposal, the Deep Sea Drilling Project asked that drill collars (size and bore) be specified and include sufficient shock bumper subs for the drill string. In addition, bumper sub size and make were requested. The shock bumper sub was identified at the Proposers' Conference as a balanced (hydraulically) bumper sub.

The proposal of Global Marine Inc. included hydraulically balanced bumper subs, but stated it was believed that the cost of maintenance would be high since this work aboard a vessel would be very difficult or impossible.

The Project made an industry search and found no balanced bumper subs with a 4 1/8inch bore had been manufactured. Such subs had been designed for Project Mohole, but never had been detailed built. A sub with a 60-inch stroke would be 30-feet long and difficult to handle.

The Deep Sea Drilling Project accepted Global Marine Inc.'s alternate proposal and the subcontract included 8 1/4-inch outside diameter by 4 1/8-inch inside diameter by 60-inch stroke Baash-Ross type OSG-2 oil packed bumper subs with 6 5/8-inch API Regular tool joints. As the 4 1/8-inch bore is not an industry standard, these subs are made on special order.

It soon became apparent that the oil-filled feature was unnecessary. As all drilling and coring is done without returns, the drilling fluid is always free of cuttings. Also, in general, formations drilled and cored in deep ocean basins are less abrasive than those drilled on land and have less tendency to sand-up and stick bumper subs. In addition, the service break is the weakest joint in the bottom hole assembly and an oil-filled sub requires two service breaks against one of a non-oil-filled bumper sub.

Baash-Ross Division, Joy Oil Tools, Houston, Texas, was the supplier of these non-oilfilled bumper subs. The male and female splines of this type bumper sub are exposed to the annulus. Vent holes are drilled in the body to allow free flow of annulus fluid as the bumper sub opens and closes. These subs were built so the female spline section is at the lower end of the male section when the bumper sub is closed. Thus, if a bumper sub is used in the closed position while drilling below the sea floor, cuttings can fill up the exposed male spline section, locking the equipment in the closed position.

There was some question whether Baash-Ross could fit the necessary number of subs into their production schedule, so the Project accepted the substitution of an equivalent design by Shaffer Oil Tool Company, of Beaumont, Texas.

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When the drill collar tool joint was changed from 6 5/8-inches API Regular to NC-61 at the end of Leg 9, change over subs were made so the bumper subs could be run inverted. This reduced the tendency for the equipment to freeze.

Concurrently with these changes, bumper sub suppliers, Global Marine Inc. and the Project were conducting studies to reduce bottom hole assembly failures.

One area of suspect was the change in impact properties and yield strength of bottom hole assembly components operating at the low temperatures (33°F) found in deep water. The International Nickel Company, Inc., of New York City, was very helpful in furnishing reports on some of their tests on properties of steels at low temperatures. These and drilling experiences in Alaska and Northern Canada indicated the 4142 steel being used should not be affected at this temperature. As 4340 steel has some properties that are superior to 4142, the bumper sub suppliers are using this steel.

Shaffer Oil Tool Company proposed a bumper sub that has the packing for the wash pipe accessible from the inside bore. This eliminates the necessity of a service break which increases the strength of the bumper sub.



The service break of a bumper sub is the weakest joint in the bottom hole assembly. These are typical failures of this joint.

Several bumper subs of this design have been placed aboard D/V Glomar Challenger and are being tested.

During these discussions, it was suggested many times that the outside diameter of the bumper sub should be increased so a stronger service joint could be used. Although this would increase bumper sub life and reduce bottom hole assembly losses, the consensus was that the increase in the diameter of a section of the drill string would increase the possibility of sticking the string. This is a very real possibility in the sites which have been drilled through alternating very hard layers and very soft layers. A high volume of drilling fluid is necessary to penetrate hard layers. This erodes pockets in the soft layers and continued drilling then fills up these pockets with cuttings which fall back when circulation is stopped. If the enlargement in the drill string is immediately below this cavity, bridging and sticking of the drill string is then probable.

Since the inception of this program, the state of the art of motion compensators has advanced rapidly.

At present, one system designed and built by VETCO Offshore Industries, Inc., of Ventura, California, is being field tested. This system uses a pair of hydraulic cylinders to connect the hook to the traveling block. The hook is lowered or raised to compensate for vessel motion. In effect, a constant pressure is maintained below the hydraulic pistons. Oil is added or bled off to compensate for vertical vessel motion. The motion sensor directs oil flow. A gas-oil accumulator evens out oil flow and reduces energy input requirement.

Field trials look very promising. Several units are being installed on floating platforms and on the SEDCO 445 drill ship.

The second system uses one large hydraulic cylinder to move the crown block sheaves up and down to compensate for ship motion. This system is being installed on a Royal Dutch Shell drill ship.

All studies todate indicate a motion compensator would eliminate the need for bumper subs and would increase drilling penetration and core recovery. The Deep Sea Drilling Project is following these units with interest. A three-year extension of operations (Phase II) would make it feasible to investigate installation of a motion compensator aboard D/V Glomar Challenger.

APPENDIX (Bumper Subs)

- 1. Bumper Sub Recommended Inspection and Service.
- 2. Charpy Impact Curves.

BUMPER SUB RECOMMENDED INSPECTION AND SERVICE



BUMPER SUB

RECOMMENDED INSPECTION AND SERVICE

TYPE GSI-PM-1 RENEWABLE SPLINE TYPE

> BAASH-ROSS DIVISION OF JOY MANUFACTURING CO.

/14 BROCHURE NO. 4-1-69

RECOMMENDED INSPECTION AND SERVICE OF 6SI-PM BUMPER SUBS

TOOL JOINTS AND ASSEMBLY THREADS

After each run, inspect the Tool Joint box and pin for erosion marks, gall marks, nicks or pulled threads. The Tool Joint shoulders should be checked for signs of fluid lapping which would indicate that the joint was not made up with sufficient torque. Characteristic of a loose joint is a dry pin and a dull gray color on the area around the joint shoulder caused by fluid lapping in and out between the shoulders.

After 10 or 12 repacks, or approximately 1000-1200 hours of running time, the tool joints should be magnafluxed for minute fatigue cracks that may have resulted from improper joint make-up. Excessive or insufficient make-up torque will eventually cause cracks to appear in the first or second threads near the base of the pin. If the tool joints are damaged to any great extent, they should be recut.

After each repack, inspect connecting or assembly threads for similar type of damage as above. The joint shoulders should also be inspected for extreme flaring or compression of metal indicating a loose connection. The fin, or flare, on the pin joint shoulder can be removed by grinding. If the assembly joints do not have the recommended draw to shoulder for power make-up, (see separate sheet), the box joint face should be faced back as required.

NOTE: Refer to recommended joint make-up torque as shown on separate sheet of this report for the specific joint and size of Bumper Sub.

BODY

After each run, the body O.D. should be checked for wear. The body should be replaced if worn 3/16" under the original diameter. The top box connection should be checked for swelling. The swelling of the box joint would indicate that the body O.D. is worn more than recommended, a loose connection has developed or that excess torque has been transmitted through the tool. If this joint does not have the recommended draw to shoulder for power make-up, the joint face should be faced back as required.

SPLINE INSERTS

For best results, the Spline Inserts should be replaced when they are worn 50% of their original width.

MANDREL

After each repack, the spline width should be checked for wear. If the spline is worn more than 50% of the original width, the Mandrel should be replaced. The amount of wear can be determined by visual inspection of the spline on the lower end of the Mandrel where wear does not occur. The bumping shoulder on the top end of the Mandrel immediately below the stuffing box should be inspected for mushrooming and should be ground flush with 0.D. before reassembling.

After 10 or 12 repacks, the lower connection should be magnafluxed for minute cracks that may have developed from extreme torque or flexing of the joint due to a loose connection. At this time, the pin joint threads should be inspected for erosion marks, gall marks, nicks, or pulled threads. If flaring or mushrooming has developed on the spline end (joint shoulder), spline should be ground as required.

After each repack, the packing cell in the top end of the Mandrel should be checked for signs of erosion or washing out of the sealing surface. The threads in the packing cell should be checked for such damage as pulled or crossed threads and/or worn threads. Such damage can be repaired by building up with weld and remachining to size.

PACKING BUSHING, FOLLOWER RING, PACKING NUT & PACKING

The Packing Bushing and Follower Ring are bored 1/32" larger than the Washpipe O.D. and for best results should be replaced if worn 1/16" or more on diameter, that is, if the I.D. is larger by 3/32" or more than the O.D. of the Washpipe. This wear tolerance would not apply to the Packing Nut as its larger bore does not effect the sealing capabilities of the packing. The Packing Nut, however, should be checked for worn, stripped, or crossed threads and replaced as required. Generally speaking, depending on the particular area and on the type of operation, the packing should last upward from 60 hours, unless the Packing Bushings or Washpipe are damaged or worn beyond the recommended tolerance.

WASHPIPE

After each repack, the hard chrome surface on the Washpipe should be checked for pits or score marks. If the Washpipe has developed a few pits but looks like it can be salvaged, brazing can be applied to the pitted area and ground flush. Exposed edges on the chrome can cause further damage to the Washpipe or to the packing. If it is required that the Washpipe be replaced due to prolonged use in salt water or other corrosive elements, it is suggested that the new Washpipe be secured in place using teflon tape on the threads and making up the joint to the recommended torque (see attached sheet).

RECOMMENDED JOINT MAKE-UP TORQUE FOR ASSEMBLY OF BUMPER SUBS

6-5/8 A.P.I. REG. 8-1/4 O.D. X 4-1/8 I.D. X 60" STROKE TYPE 6SI-PM BUMPER SUB (6 SPLINE INSERTS) COMPLETE PT. NO. 7521-1212-5

CONNECTION	TORQUE FT./LBS.
6-5/8 A.P.I. Reg. Tool Joints (4-1/8 I.D.)	33,000 - 36,000
Top Sub to Body Thread (1/16-1/8 Draw to Shoulder)	38,000 - 42,000
Washpipe to Top Sub Thread	2,500 - 3,000
Mandrel to Bottom Sub Thread (1/16-1/8 Draw to Shoulder)	16,000 - 18,000

NOTE: To prevent galling of joints at make-up and in service due to high torsional loads, all threads and shoulders should be lubricated with a good quality lubricant recommended for such service. The lubricant recommended for this type of service is one that will provide sufficiently high coefficient of friction to prevent further make-up of joint in transmission of torque.

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BAASH-ROSS DIVISION OF JOY MANUFACTURING COMPANY

PARTS LIST

DESCRIPTION

BUMPER SUB 6SI-PM-1 RENEWABLE SPLINE TYPE 8-1/4 O.D. X 4-1/8 I.D. X 60" STROKE 6-5/8 A.P.I. REGULAR TOOL JOINT. COMPLETE PART NO. 7521-1212-5

NO. REQ'D.	PART NO.	NAME OF PART
1	7521-A-1058	BODY
1	7521-A-1054	MANDREL
1	7521-1212-C	"O" RING (MANDREL)
1	7521-A-1059	BOTTOM SUB (PIN)
1	7521-A-1060	TOP SUB (BOX)
1	7521-1212-A	"O" RING (TOP SUB)
1	7521-A-1055 GR	WASHPIPE
1	7521-S-42	LOCK RING
1	7521-B-275	PACKING NUT
1 SET	7521-A-266	SPLINE INSERT (6 INSERTS)
1	7521-C-1057	FOLLOWER RING
1	7521-A-1057	HIGH PRESSURE PACKING BUSHING
1	7521-B-1057	LOW PRESSURE PACKING BUSHING
2	7521-1212-В	"O" RING (PACKING BUSHING)
1	7521-L-924	SEAL RING
1	7521-1212-D	"O" RING FOR SEAL RING
1 SET	7521-1212-E	LOW PRESSURE PACKING
1 SET	7521-1212-F	HIGH PRESSURE PACKING
1	7521-A-1056	WRENCH
1	7521-A-32	LOCK RING INSERTER
1	7521-A-33	LOCK RING GROOVE CLEANER
1	7521-A-34	LOCK RING REMOVER







CHARPY IMPACT CURVES

Excerpt from report: Notch-Impact Transition Temperature of Quenched and Tempered Low Alloy Steels. F.L. LaQue. W.A. Wesley. June 29, 1956. The International Nickel Company.





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INTERNATIONAL NICKEL








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DRILL COLLARS

DRILL COLLARS

In response to the Deep Sea Drilling Project Request For Proposal, Global Marine Inc. proposed 8 1/4-inch outside diameter by 4 1/16-inch inside diameter by 30-foot drill collars with 6 5/8-inch API Regular tool joints. The outside diameter of the collar would match the outside diameter of the proposed bumper subs, and in a 9 7/8-inch hole, would leave adequate hole clearance to permit free circulation. The 4 1/16inch bore would give clearance for the wireline retrievable core barrel. In addition, 7 1/4-inch outside diameter by 4 1/16-inch inside diameter by 30-feet drill collars with 5 1/2-inch API Full Hole tool joints were to be used as transition joints between the 8 1/4-inch collars and the drill string.

These same collar specifications were carried over to the subcontract. The collars were actually bored 4 1/8-inch inside diameter to match the inside diameter of the drill string tool joint. In addition, 5 1/2-inch range three heavy wall drill pipe was to be placed aboard D/V Glomar Challenger to assist in the transition from the 7 1/4-inch outside diameter collars to the 5-inch drill pipe.

A 6 5/8-inch API Regular tool joint is normal on 8 1/4-inch outside diameter collars. However, the increase in bore from the normal 2 13/16-inch inside diameter to 4 1/8inch inside diameter gives and unbalanced joint in that the cross section of the pin is reduced and is much weaker than the box. This tool joint was approved by the Deep Sea Drilling Project and Global Marine Inc. as it is a universal joint with gauges and hobbs available in many countries for repair. Also, the initial Program Plan envisioned drilling and coring soft sediments with minimum recovery of basement and hard formations.

It soon became apparent that chert and other hard formations could be and were being encountered well above basement. Drill collar tool joint and bumper sub service joint failure increased when attempts were made to penetrate these layers as the core bits available were mostly diamond or sintered carbide drag bits. Failure because of high downhole torque seemed to predominate. Excursions of the ship because of erratic positioning before the drill collars were drilled below the mud line also increased losses.

In areas such as the Northwest Pacific where resistant horizons were encountered at shallow depths beneath the bottom, insufficient soft overburden existed to properly stabilize the bottom hole assembly. After a number of failures associated with bumper subs or drill collar joints, it was concluded that these failures resulted from excessive working of the unsupported and relatively inflexible components of the bottom hole assembly.

Initially upon this determination, an operational restriction was placed on the drilling activities of D/V Glomar Challenger requiring a minimum of 100 meters of generally unconsolidated overburden above the first resistant horizon.



Typical drill collar tool joint failures. These failures have been reduced by changing to a balanced tool joint.

Satisfactory efforts to increase the strength of bottom hole assembly components have led to a relaxation of this operational limitation and during Phase II, 50 meters has been regarded as the minimum permissable thickness of uppermost unconsolidated material.

In at least one site by carefully entering by means of minimum bit weight, penetration of a resistant layer at approximately 20 meters of depth was safely and satisfactorily accomplished.

Before the restriction, bottom hole assembly losses were twice the projected rate and it was necessary to air lift drill collars, core barrels and bumper subs to San Juan, Puerto Rico, where D/V Glomar Challenger made an unscheduled stop between Sites 27 and 28 on Leg 4 to pick them up.

The Deep Sea Drilling Project contacted oil field tubular goods experts among the major oil companys and leading contractors and finally engaged the services of a consultant, Mr. W.S. Crake (Chairman, API Tubular Goods Committee, Gulf Coast Division) to review the drill string design.

A review of bottom hole assembly losses and industry comments in addition to those of the Deep Sea Drilling Project, Global Marine Inc., and suppliers resulted in the following decision: Resupply D/V Glomar Challenger with the same bottom hole assemblies as used on the first four legs. This decision was based on several factors:

- Although bottom hole assembly loss was high, total footage drilled, footage cored and footage of core recovered was far above projected rates, indicating a balanced assembly.
- (2) Lead time before entering the Pacific was very short to change drill collar, bumper sub and bit size.
- (3) Drilling conditions in the Pacific were unknown.

The track in the Pacific originally was scheduled to go from the Panama Canal to Tahiti, to Honolulu, to Guam, to Honolulu and into Los Angeles. This track was changed to have Leg 4 end in San Diego, and Leg 5 go from San Diego to Honolulu, or in the exact reverse of the original schedule. The port call at San Diego allowed resupplying in a stateside port and for maintenance on drill collars and bumper subs in shops knowledgeable of this type of equipment.

A minimum of chert or hard formations was encountered on Leg 5. However, on Leg 6 and following legs, many chert and hard formation layers were found at shallow formation depths, indicating the condition would be prevalent in the Pacific.

Concurrent with this discovery, the Deep Sea Drilling Project and Global Marine Inc. were conducting an industry search to determine the best tool joint 8 1/4-inch outside diameter and 4 1/8-inch inside diameter. Both the 5 1/4-inch API Internal Flush and the NC-61 were acceptable. Global Marine Inc. felt more comfortable with the NC-61, so it was established as the tool joint for 8 1/4-inch outside diameter by 4 1/8inch inside diameter tubular goods. Before the end of Phase I of the Project, it was planned to re-cut all drill collars, bumper subs, change over subs and other allied equipment to this thread at the first opportunity.

This opportunity came between Legs 9 and 10 when D/V Glomar Challenger went into drydock at the Tood Shipyards in Galveston, Texas. The numerous drill collar manufacturers in nearby Houston made it possible to re-cut all drill collars and bumper subs to the NC-61 thread.

Since this change over, the Deep Sea Drilling Project has not had a failure in an NC-61 tool joint. This, in part, is probably because of the increased use of carbide insert roller core bits. Roller bits are less likely to subject the bottom hole assembly to the high instantaneous torques caused by drag bits.

As drilling rates, coring rates and core recovery remain high, there are no plans to change drill collar specifications.

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CORING EQUIPMENT

A request to include core barrels and accessory items was contained in the Deep Sea Drilling Project's Request For Proposal which went to bidders in 1967. It also specified continuous coring with wireline core barrels using commercially available equipment. Bidders were told clear plastic liner tubes could be used in the inner barrel.

In their answer, Global Marine Inc. proposed to furnish a Christensen 6 3/4-inch outside diameter core barrel to take a 2 1/2-inch core. This barrel would be equipped with center drills, mechanical punch corers and piston cores. It could also be used to take oriented cores and several types of catchers were available.

After evaluating all proposals, the Deep Sea Drilling Project decided it would be to the advantage of the Project if the coring equipment was selected and supplied by the Project.

A statement of work was written and incorporated in a Request For Quotation (RFQ). This RFQ was sent to all suppliers of oil industry coring equipment and two companies, Christensen Diamond Products Company, of Salt Lake City, Utah, and Hycalog Inc., of Lafayette, Louisiana, were responsive. Both companies were very active in coring – including wireline coring.

Two different types of coring equipment were proposed. Christensen proposed a version of the Reed Robishaw wireline retrievable barrel. This barrel has the landing collar at the upper end of the inner barrel. A spring-loaded latch is placed immediately above this collar to hold the inner barrel down for retrieval. A bearing hanger allows the inner barrel to remain stationary while the core is being cut. Christensen also proposed extending the inner core barrel below the core bit to prevent washing the core. The barrel would be spring loaded to retract flush with the core bit. The core would be 2 1/2-inches outside diameter, requiring a bit with a 3 1/2-inch center opening. This barrel also could be made up solid on the center bit assembly to cut punch cores, while a solid center bit would be used when drilling ahead.

The Hycalog design used an inner barrel with the landing collar close to the bit. The landing seat is actually the inner race of a ball bearing. This bearing, in addition to stopping the inner barrel, centered the barrel to the core bit and prevented the inner barrel from drooping if the hole was at an angle. This design also had a spring-loaded latch to hold the inner barrel down and drive the center bit. The Hycalog outer barrel had a large inside diameter of 6 5/8-inches for high rates of circulation when drilling hard formations and required a core barrel thread for connections. This connection is similar to the one used on wash-over pipe and is much weaker than the drill collar connections. The RFQ asked for a 30-foot long core barrel made up in 15-foot units, as it was not known if 30-foot cores could be consistently recovered.





The Christensen-style core barrel offered was the old standby of the oil fields. However, the Hycalog barrel was designed specifically to overcome problems encountered in exploratory coring in the Gulf of Mexico. The Hycalog unit gave excellent recovery and they were used for the Blake Plateau Coring Program during April-May of 1965. This program was sponsored by the Joint Oceanographic Insitutions for Deep Earth Sampling (JOIDES) and funded by the National Science Foundation.



The original core barrel had a 9 1/4-inch outside diameter bit and was run with a stabalizer immediately above the bit.

Since there was insufficient time to test coring equipment and select a supplier for Leg 1, the Deep Sea Drilling Project had both Christensen and Hycalog core barrels aboard D/V Glomar Challenger when the Project became operational on August 11, 1968.

Since Leg 1 would be in the Gulf of Mexico, the Project decided to use the Hycalog barrel as the primary equipment with two Christensen barrels aboard for evaluation.

It soon became apparent that the Hycalog connection was too weak. Formations encountered were much harder than expected. A review of drilling and coring on Leg 1 indicated much higher circulating rates were not required, so the Hycalog bit, head, and top bumper subs for the outer barrel were changed so that a standard 30-foot drill collar could be used as an outer core barrel.

Several units of core barrel conversion subs to adapt the Hycalog equipment to run with a standard drill collar were placed aboard D/V Glomar Challenger at San Juan, Puerto Rico, during Leg 4. This configuration proved to be an improvement and is now the accepted core barrel design. The tool joints to the drill collar were changed to NC-61 when the collars were re-cut at Galveston. Core recovery with the 30-foot assembly was good, so eliminating the 15-foot barrel did not reduce core recovery.



Core barrel latch and Reed PDD spear point.

The first core barrels of both Hycalog and Christensen used the Reed PDD spear point on the latch pull rod and the Reed PDD overshot to retrieve the filled core barrel. The crews experienced excessive mechanical failure due to the large drill collar bore since PDD equipment is normally run in 3 3/16-inch inside diameter drill collars. Conversion units to a 3-inch Otis fish neck spear were made and could be attached to the pull rod and an Otis latch used for retrieving. This proved a more dependable unit so replacement latches were built with Otis spear points.

Other weak points in the coring equipment have been modified or made stronger.



Plastic core liner collapsed and wadded up into the inner core barrel.

The clear buterate plastic liner has been a problem on several occasions. In addition to collapsing, the liner would wad inside the barrel. This was eliminated by reinforcing the lower or entrance end of the tube with a metal sleeve.

Many different sediments were drilled and cored during the first three years of operation. In general, these sediments have been soft and unconsolidated or very hard such as chert or limestone. Since both types were often encountered in the same hole, it has been difficult to build bits and core catchers that perform satisfactorily in such extremes. Some of the core bits used on Deep Sea Drilling Project:



Massive set diamond bit with center bit.



Cross section milled tooth roller bit.

Three cutter carbide insert roller bit.

The tungsten carbide insert roller bit has turned in the best performance todate. (See upcoming Technical Report on Core Bits.)

Double dog-type core catchers are run more than any other combination. The inner core barrel is designed to run tandem catchers. Normally, a dog catcher is run above a hard formation collet catcher. However, the Deep Sea Drilling Project coring is done without stabilization of the bottom hole assembly. This increases the percent of hard formation core cut under gauge so the collet catcher can not hold it.



Dog type core catcher in place in the inner core barrel.



Core catchers used with wireline retrievable core barrel on Deep Sea Drilling Project.

During Leg 2, Dr. Melvin N.A. Peterson, the Project's Co-Principal Investigator who was a cruise co-chief scientist on the New York to Dakar expedition, added a pliofilm tubular catcher above the upper dog catcher. This allows sampling of very soupy sediments and has even held sea water.

The drilling crews have been constantly speeding up and improving coring and drilling operations. Coring technicians from oil well coring companies were used to supervise coring on the first nine legs. At this time, the Project thought the drilling crews were familar with the techniques necessary for maximum coring recovery, so the coring technician was replaced by adding a Global Marine Inc. crew member. His duties were to assist in coring, maintain the coring and other special equipment furnished by the Deep Sea Drilling Project.

Drilling and coring operations have followed rather closely to the procedures outlined in the Program Plan. The Project shipboard complement includes a cruise operations manager. This person may be a University of California at San Diego employee, or may be on loan from the Drilling Department of a major oil company. Either way, he is the Deep Sea Drilling Project's representative. Since these men have varied backgrounds, drilling techniques may vary. However, unless unusually adverse drilling conditions are encountered, core recovery is consistent.

The use of drilling mud has varied from leg to leg. Since there are no mud returns to

D/V Glomar Challenger, it is difficult to evaluate what effect the mud has in cleaning the cuttings out of the hole.

Currently, the Deep Sea Drilling Project is engaged in designing and testing the following:

- 1. A piston core barrel for mud line cores.
- 2. Extended inner core barrel for less-disturbed cores.
- 3. Pressure core barrel for in situ sampling.

The results of these efforts will be subjects of later technical reports.

See Bibliographies on Drill Pipe, Drill Collars and Bumper Subs.

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DEPTH IN FEET

Water Total Depth Penetration Site (ft.) (ft.)		String Length (ft.)	Type of Failure*	Remarks	
Acceptance					August 1968
Trials	3,584	60	3,644	(Bending) upper joint core barrel	off station Leg 2
8	17,006	847	17,853	upper joint core barrel	Hycalog thread
12	14,936	220	15,156	wobbled off second drill collar	off station Leg 3–January 1969
20	14,779	82	14,851	service joint bumper sub	high torque observed Leg 4
24	16,924	770	17,694	est. service joint bumper sub	failure point dropped with string
24	16,924		-	snapped off 2 feet below box 479 joints off bottom	picked up elevators with one link
25	6,355	252	6,607	service joint bumper sub	high torque
26	17,014	60	17,074	tool joint between 1 and 2 drill collar	B.H.A. not buried
26	17,014	1,958	18,972	upper joint core barrel	Hycalog thread Leg 5
35	11,063	1,279	12,332	service joint top bumper sub	(after logging) Leg 6
44	4,872	269	5,131	wobbled off top of sixth drill collar	pin bent
45	18,105	100	18,205	service joint on bumper sub	
45	18,105	343	18,448	torsion failure sixth joint above drill collar	no bumper subs
46	18,960	70	19,030	pin first drill collar above bumper sub	bumper sub above con barrel
58	14,680	560	15,240	parted by string shot middle bumper sub	drill string stuck Leg 8
68	17,936	49	17,975	pin third drill collar above core barrel	shallow chert

DRILL STRING LOSS THROUGH LEG 18

*Resulting in loss of equipment below the keel.

Drill String Loss Through Leg 18 concluded.

	Water Depth	Total Penetration	String Length	T	D
Site	(ft.)	(ft.)	(ft.)	Type of Failure	Remarks
					Leg 9-January 1970
76	15,085	90	15,175	tool joint second drill collar above core barrel NOTE – Changed drill collar thread to NC61	shallow chert
					Leg 10
87	12,350	2,285	14,635	backed off 29 stands above drill collar	drill string stuck
93	10,015	5	10,020	top service break bottom bumper sub	lost beacon large excursion
93A	10,040	26	10,066	top service break top bumper sub	shailow clay Leg 11 A,B,C
99	16,123	1,276	16,399	top service break second bumper sub	tight hole
99A	16,123	814	16,937	top service break second bumper sub	firm formation
100	17,471	1,086	18,557	top sub bottom bumper sub	cracked box (old crack)
106	14,764	1,184	15,948	top service break middle bumper sub	high current erratic positioning
109	10,000	-	-	re-entry equipment	see re-entry report Leg 14
143C	10,020	160	10,180	top service break bottom bumper sub	sloping bottom
143D	10,020	59	10,079	top service break bottom bumper sub	sloping bottom Leg 15–January 1971
146	13,300	_	1.5	re-entry equipment	see re-entry report Leg 16
163	17,500	958	18,458	re-entry equipment	see re-entry report Leg 17
168	17,800	246	18,046	sub 6 5/8 double box top bottom bumper sub	
					Leg 18
176	630	134	764	service joint bumper sub	50 foot excursion – shallow water
177A	7,000	1,670	8,670	bumper sub packing nut backed off	test bumper sub

			Average No.								
eg				Water Depth	Water Penetration		of	Footage	Footage	Total	Percen
Vo.	Dates	Holes	Sites	LO - High	Depth	L0 - High		Cored		Penetration	
	8 Months:										
	8/11/68 - 9/23/68	11	7	9,259 - 17,567	15,539	78 - 2,528	55	1,411	595	10,432	42
	9/30/68 - 11/25/68	10	5	11,716 - 17,006	15,038	76 - 2,738	78	2,104	713	9,955	34
1	12/ 1/68 - 1/24/69	17	10	6,928 - 15,337	12,873	68 - 1,591	102	2,697	2,214	11,831	90
V	1/27/69 - 3/24/69	15	9	3,994 - 18,109	13,553	215 - 1,832	120	3,536	1,217	13,574	34
1	4/12/69 - 6/ 5/69	13	12	10,735 - 17,728	15,234	30 - 1,279	131	3,355	2,854	6,260	85
/1	6/12/69 - 8/ 5/69	34	17	4,848 - 19,614	13,637	33 - 1,151	124	3,211	2,274	8,324	71
/11	8/ 8/69 - 10/ 3/69	15	7	6,758 - 20,146	14,392	15 - 3,231	146	3,865	3,078	15,574	80
/111	10/ 8/69 - 12/ 2/69	8	8	13,749 - 17,969	15,304	79 - 1,830	181	4,842	4,019	6,642	83
Х	12/ 6/69 - 1/31/70	17	9	10,159 - 15,115	13,508	30 - 1,579	202	5,554	5,045	12,331	91
	Sub-Totals	140	84				1,139	30,575	22,009	93,923	72
xtens	ion 30 Months:			2001 2001							
<	2/13/70 - 4/ 5/70	14	13	4,859 - 12,346	9,474		162	3,932	2,421	20,405	62
<1	4/ 8/70 - 6/ 1/70	15	11	6,086 - 17,504	12,931	250 - 3,334	153	3,936	2,127	18,815	54
e-En	try Trials:										
	6/ 3/70 - 6/16/70	2	2	9,941 - 10,017	9,979	207 - 292	3	103	73	499	71
(11)	6/19/70 - 8/11/70	13	9	3,438 - 16,113	9,301	330 - 2,831	184	4,754	2,745	19,381	58 ,
KIII	8/14/70 - 10/ 6/70	28	15	3,849 - 15,303	9,899	35 - 3,170	201	4,733	2,176	20,941	46
XIV.	10/ 9/70 - 12/ 1/70	18	10	9,735 - 17,622	13,332	59 - 2,260	. 98	2,495	1,341	16,728	54
XV	12/3/70 - 2/2/71	16	10	2,927 - 14,912	10,544	74 - 2,594	264	8,042	4,120	15,072	51
XVI	2/ 5/71 - 3/30/71	12	9	6,408 - 17,455	12,865	13 - 1,759	210	5,781	4,166	8,266	72
XVII	4/ 4/71 - 5/25/71	9	8	7,530 - 19,004	15,435	246 - 3,888	-237	7,511	2,938	10,618	39
XVIII	5/28/71 - 7/19/71	15	11	633 - 16,152	9,190	75 - 2,884	254	8,191	4,003	12,793	49
<ix< td=""><td>7/22/71 - 9/12/71</td><td>16</td><td>11</td><td>6,300 - 15,818</td><td>11,876</td><td>30 - 3,468</td><td>-255</td><td>7,205</td><td>3,402</td><td>32,692</td><td>47</td></ix<>	7/22/71 - 9/12/71	16	11	6,300 - 15,818	11,876	30 - 3,468	-255	7,205	3,402	32,692	47
<x td="" <=""><td>9/17/71 - 11/10/71</td><td>13</td><td>9</td><td>4,853 - 20,323</td><td></td><td>217 - 1,499</td><td>30</td><td>1,591</td><td>531</td><td>9,715</td><td>34</td></x>	9/17/71 - 11/10/71	13	9	4,853 - 20,323		217 - 1,499	30	1,591	531	9,715	34
(XI	11/16/71 - 1/11/72	13	8	4,590 - 15,266	9,959	154 - 2,408	283	8,459	4,551	15,444	54
	Sub-Totals	184	126	6. J. C. C.			2,339	66,733	34,594	201,369	52
	Grand Totals	324	· 210				3,478	97.,308	56,603	295,292	58

DEEP SEA DRILLING PROJECT

NOTE: All measurements are in feet.

Fedruary 1972

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APPENDIX A

Metal Corrosion in Saline Waters

DISTRIBUTION OF SALINE WATERS IN THE UNITED STATES



METAL CORROSION IN SALINE WATERS

The corrosion problems inherent in the handling and use of saline waters can be costly. The author discusses some of the mechanisms of such corrosion and suggests some steps for countering saline-water attack on metals.

by Frederick W. Fink

IN MOST AREAS OF THE United States, as of many countries, problems associated with obtaining an adequate supply of fresh water are multiplying. Such conditions as rising urban populations, pollution of available supplies, and demands of agriculture and industry are forcing the exploitation of saline waters by a growing number of individuals, governments, and other organizations.

Saline waters, because of their wide distribution, are increasingly being considered for possible use in applications such as industrial cooling, fire protection services, street working and sanitation, and a rawwater source for conversion to potable water.

Waters containing 1000 ppm or more of salt are usually referred to as saline or brackish waters. Such waters are available in many parts of the USA, sometimes as surface water and often from wells.¹ In a few cases, wells produce brines that have industrial values as sources of chemicals. In the future, however, even fresh water sources can be expected to have increasing salt contents in some areas as a result of such developments as (1) greater pollution or water reuse, (2) use of salt for highway ice and dust control, (3) excessive pumping of wells (which in coastal areas allows infiltration of sea water), and (4) evaporation from large reservoirs, leaving the salts behind.

Saline waters, as a natural resource, pose many difficulties. Plants using saline water for cooling or as a source for conversion to fresh water, for example, have the problem of disposing of the salt-bearing effluent. In oil-producing fields the connate brines produced are disposed of by pumping them into a formation which precludes contamination of other underground values that may be present.²

The Office of Saline Water in Washington is conducting a wide range of experimental programs on methods of converting salt-bearing waters to fresh. It is confident that fresh water eventually can be produced from saline sources for well under \$1.00 per thousand gallons. Nevertheless, the economics of saline



Corrosion increases with rising salt concentration. However, at some point (3 to 4 per cent for sodium chloride), the corrosion rate drops because oxygen is less soluble in the solution.

waters involve excessive expense when compared with fresh water. There are higher costs of handling, processing, and finally disposing of the salt residues. Not the least of these higher costs result from the steps needed to prevent excessive corrosion of pipes and equipment.

FACTORS IN CORROSION BY SALINE WATERS

Saline waters usually are more corrosive to metals than ordinary fresh waters.³ To predict whether saline well water will be corrosive to steel, it is necessary to understand the effects of such factors as (1) the salt content, (2) the dissolved gases, (3) the pH, (4) the temperature, and (5) the tendency to form mineral scale.⁴ If the water contains several thousand parts per million of sodium and other chlorides, it is likely to be corrosive to steel and low-alloy steels. Stainless steels, copper alloys, aluminum alloys, and some nickelbase alloys also may be attacked, depending on conditions.

Salt Content. Distilled water in contact with air is corrosive to steel, probably because, in addition to oxy-



Backed by a broad knowledge of the properties of materials and of the characteristics of various environments, Frederick W. Fink has participated in and guided a broad spectrum of corrosion studies. Two of his current interests include studies involving the oil-ash problem in boilers and the deterioration of materials in marine service. In recent years, he has contributed to research on the mechanism of corrosion cracking in austenitic stainless steel, acid corrosion of cast stainless steel, and corrosion in nuclear reactors and

corrosion in nuclear reactors and rocket propellant systems. His interests at Battelle have included oil-well equipment corrosion, effects of flue-gas on metals, and development of alloys for chemical service. An important contributor to the work of professional societies, Fink has long been active in organizing the corrosion symposia of the National Association of Corrosion Engineers and The Electrochemical Society. gen, carbon dioxide is present which makes it slightly acid. If a small amount of sodium chloride is added to such water, the corrosivity to steel is increased greatly. In fact, as sodium chloride is added, the corrosion rate increases up to a point at which a concentration of about 3 or 4 per cent is reached. Beyond this point, the corrosion rate starts to decrease (see figure at left). Thus, in a concentrated brine the corrosion rate is less than in, say a dilute salt solution, because of the much lower solubility of oxygen in the more concentrated solution.

After the chloride ion, the sulfate ion is most corrosive to steel. One explanation for the chloride ion's corrosiveness is that its smaller size enables it to penetrate protective films more readily than sulfate ions. Frequently, protective films form naturally on the surface of steel or other metals, or are developed by inhibitor additions. One can consider sodium and potassium ions as being corrosive, because they do not form protective mineral scales on the metal surface. On the other hand, calcium and magnesium ions are well known for their tendency to precipitate protective mineral scale on heated metal surfaces.

Corrosion Cell Mechanism. The attack on a metal such as steel immersed in a saline water is electrochemical in nature. Electric currents can be detected entering or leaving the solution at local sites in a corroding structure. Metal ions, e.g., iron, enter into solution at certain active areas referred to as "local anodes." The other electrodes of the local corrosion cells are known as "cathodes," where certain ions, usually hydrogen, tend to deposit from solution. Any process that interferes with the reaction going on either at the local anodes or cathodes tends to restrict the over-all operation of the corrosion cells.

Role of Oxygen and Other Gases. Corrosive attack on steel in saline well waters having a pH of about 5 to 9, or higher, is usually controlled by the rate that oxygen, in solution, diffuses to the cathode. In the corrosion process, oxygen will remove the deposit of hydrogen at the local cathodes by forming water. In a flowing system containing plenty of oxygen, the cathodic areas receive an ample supply, and the local corrosion cells can continue to function at a fairly rapid rate. If oxygen or some other oxidizing medium is not present, the hydrogen film tends to accumulate on the cathode surfaces and the corrosion cells cease to function.

At the anode where the iron enters solution as ferrous ions, the presence of oxygen under some circumstances may promote protective oxide-film formation. Oxygen also can cause ferrous ions to oxidize to ferric ions. The latter can, in turn, attack a fresh steel surface. In general, oxygen is considered the most important of the environmental factors in saline waters. In a saline water open to the atmosphere, oxygen is normally present up to 8 ppm in solution at ordinary temperatures. If pressure is applied to the water, much larger amounts of oxygen can be maintained in solution.

In addition to carbon dioxide and oxygen, hydrogen sulfide can promote corrosion when present in saline water, even in small amounts. For example, as little as one ppm greatly accelerates the corrosive attack on steel. When hydrogen sulfide is present in a well water, it is often the result of bacterial action. Hydrogen sulfide is producing by sulfate-reducing bacteria and by bacteria which promote decomposition of organic sulfides.

Some connate brines, originating deep down in the earth and under high pressure, are found to have no oxygen present. If the pH is 6 or above, these oxygenfree brines are not corrosive to steel, even though the temperature may be as high as 250 F. Occasionally a connate brine is found which not only is oxygen-free



In saline waters above 5 pH, the greater the amount of oxygen diffusing to cathodic areas, the more tendency for corrosion to proceed.

but will also chemically react with small amounts of oxygen (air) that may be inadvertently introduced during handling. In other cases it is found desirable, when handling connate brines in steel piping, to add a scavenger of oxygen (such as sodium sulfite) to maintain the water oxygen-free and noncorrosive to steel.

Acid. Saline waters may be contaminated by mine water or other sources of acid. If the pH is much below 5, oxygen is not needed for corrosion of steel. For example, a direct reaction will occur between the metal and the solution, liberating hydrogen.

Temperature. Increasing the temperature of a saline water tends to accelerate its attack. At atmospheric pressure, however, oxygen solubility decreases as the temperature is raised, and at some point this becomes the dominating corrosion factor. For example, boiling saline water is normally less corrosive than water at some lower temperature, say 160 to 180 F.



Saline solutions tend to become more corrosive with increasing temperature. At atmospheric pressure, however, oxygen solubility decreases with increasing temperature, and near and at the boiling point the solution's corrosivity is less than at lower temperatures.

Velocity and Cavitation. Any occurrence in a saline water which tends to break down the protective film on the metal is considered undesirable. Film breakdown may result from high velocity. Impingement is one type of velocity attack. Bubbles of air or other gas may be suspended in the water, carried along by the stream, and impinged at the metal surface. Turbulent flow also tends to promote removal of the protective film and allows local attack to occur. For steel in a dilute salt solution under ambient conditions, the increase in corrosion with velocity is shown diagrammatically in the graph below.

Another type of attack (also related to velocity) is cavitation. In this case, at some point on the metal surface, the velocity reduces the pressure locally so that the water boils. At some other point further downstream, these vapor bubbles suddenly collapse. They may do so with a tremendous shock and actually cause loss of metal at local sites after repeated hammering. Impellers in pumps driven at higher-than-rated speed will often fail from cavitation attack. A cavitated impeller is extremely rough and often looks as if it had been eroded away. When a fresh metal surface is exposed by cavitation, considerable corrosion also may take place, especially in saline waters.



In general, the corrosion of steel tends to increase with the rate of flow of saline water at the metal surface.

Deposit Attack. If in a stagnant solution, a uniform deposit of silt covers the entire surface of the metal, it may be protected against corrosion. Normally, only a portion of the metal surface is covered by a deposit and the remainder is exposed. Under these conditions, a rapid corrosive attack will occur under the deposit because of what is known as a "differential aeration cell." Corrosive attack often proceeds under deposits closest to the edge. Under a deposit, the protective film (which may have formed on the metal) is not repaired because it is shielded from the oxygen. It becomes the anode in the corrosion cell. Adjacent to the deposit, oxygen is available. This area usually will act as the cathode. Corrosion cells associated with deposits or local corrosion products will frequently result in pitting. Sometimes the pitting attack is cyclic, changing from active to passive and vice versa. Layers of rust, mineral scale, and silt may form a tubercle.

IDENTIFYING THE CAUSE OF CORROSION

In view of the discussion so far, it is obvious that many factors enter into corrosion problems associated with well waters. To diagnose the problem and establish possible causes, several steps may be taken.

Experience has shown that examination of the well water at a laboratory distant from the source usually is not advisable. Bacteria, gas content, ferrous:ferric-ion ratio, and other such variables tend to change rapidly during transport. At least part of the analysis should be made at the location of the well, if possible. In any case, the sampling should be conducted by a person entirely familiar with both the corrosion and analytical problems. A chemical analysis of the brine usually is not sufficient to enable one to predict whether or not the water will be corrosive to the structure.

Similarly, the examination of a piece of rusted-out pipe in the laboratory usually does not allow conclusions to be made as to the cause of the attack. Both the environment and the metal structure must be studied together. By metallographic study, one can establish whether there is more than one phase in the metallic structure. Also, one can establish if there are minor harmful constituents in the metal, such as sulfur inclusions. Inclusions or contaminants in the metal can be the cause of local corrosion, but more often the environment holds the clue to the difficulty.

Some twenty years ago, in Battelle-conducted corrosion experiments, studies were made of samples of distillate and brine from a series of wells. In one test, the brine from a corrosive well and the distillate from a well in which corrosion was not serious were combined. It was found that the corrosion of the steel coupons by the brine was inhibited by the presence of the distillate from a noncorrosive well in the field. Further studies demonstrated that a corrosion inhibitor was naturally present in the distillate. On the other hand, if one took a brine from a noncorrosive well and combined it with a distillate from a corrosive well, the steel coupons would be badly corroded. The presence or absence of a natural inhibitor in the distillate was the principal factor in determining whether such wells would be corrosive to steel. Later experiments showed that the attack by a corrosive brine could be controlled by adding a suitable organic inhibitor to the well.

As our uses of saline waters increase, efforts to protect the equipment will have to be stepped up so as to make utilization of such waters economic. Based on knowledge of the mechanisms of corrosion and what we know of the effects of saline waters on various materials, designers can do much to limit corrosion attack and thus reduce the costs of handling saline waters.

In addition to the use of naturally corrosion-resistant materials, designers have access to other protective techniques. Thus, in some cases cathodic protection, protective coatings, or sacrificial metallic coatings can be utilized. Corrosion often can be controlled, too, through adjustment of the environment as by eliminating oxygen, carbon dioxide, and hydrogen sulfide; using inhibitors; or adjusting the pH of the waters.

Much, perhaps most, of today's research on corrosion is concerned with the correction of conditions that have arisen in existing structures. While water treatment, for example, is valuable in reducing costs of operations, the design of new installations to minimize corrosion is highly desirable. Because of the almost limitless combinations of factors that can play a part in corrosion problems, their solutions call for study by engineers and scientists experienced in the subject.

In the field of corrosion, as in others, continuing research is adding to the total of our knowledge. As our understanding of the mechanism of corrosion increases, new approaches to the solution of corrosion problems will emerge.

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APPENDIX B

Evaluation of Methods to Alleviate Corrosion Fatigue in Type 135 Drill-Pipe Steel for Offshore-Drilling Applications

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Evaluation of Methods to Alleviate Corrosion Fatigue in Type 135 Drill-Pipe Steel for Offshore-Drilling Applications

REFERENCE: Pettit, D. E., Hoeppner, D. W., and Hyler, W. S., "Evaluation of Methods to Alleviate Corrosion Fatigue in Type 135 Drill-Pipe Steel for Offshore-Drilling Applications," *Effects of Environment and Complex Load History on Fatigue Life, ASTM STP 462*, American Society for Testing and Materials, 1970, pp. 241–257.

ABSTRACT: Various methods of modifying the drilling environment and changing the response of the material to the environment were examined as potential methods of alleviating corrosion fatigue in Type 135 drill-pipe steel for use in offshore-drilling applications. The fatigue behavior of Type 135 steel specimens in seawater and in air was compared to the results obtained in deacrated seawater. freshwater and seawater drilling mud, inhibited drilling mud, and drilling mud with an elevated pH. No improvement in fatigue life resulted for any of the modified environments examined, several actually proving detrimental. Cathodicprotection studies were then conducted on specimens subjected to various applied current densities during fatigue cycling in seawater and a range of current densities that improved the fatigue behavior determined. The half-cell potential was determined for the steel in seawater and zinc selected as the sacrificial metal coating. Zinc-electroplated specimens fatigue tested in seawater showed a marked increase in fatigue life compared to that of uncoated specimens in seawater. Bare regions up to 3, in, wide were tolerated with no loss of protection and regions as large as 34 in, wide still resulted in an improved fatigue-corrosion behavior.

Thus, zinc electroplating of Type 135 drill-pipe steel resulted in a substantial improvement in fatigue behavior in a seawater environment. Alteration of the chemical environment resulted in no improvement of the fatigue behavior of uncoated specimens over that obtained in plain seawater.

KEY WORDS: fatigue, corrosion fatigue, testing, corrosion protection, steel, electroplating, surface coating, cathodic protection, inhibitors, drilling, seawater

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Introduction

In 1962, the National Science Foundation undertook a program designed to develop the equipment necessary for drilling a hole through the earth's crust and into the earth's mantle. The necessary equipment to accomplish this task was to be a refinement of the present-day offshore-drilling equipment used in oil fields.

One of the main considerations in any offshore-drilling operation is the selection of a material suitable for use in the drill-pipe string. Early test results revealed that one of the main limitations of pipe performance was the corrosion-fatigue behavior of the material. While it is recognized that corrosion-fatigue tests conducted under laboratory conditions are not directly applicable to the design function, they are useful in providing information regarding a relative order of merit for various approaches. The determination of the relative effectiveness of several different methods of preventing corrosion-fatigue in drill pipe is the subject of this paper.

Conceptually, there are three basic approaches to the prevention of corrosion fatigue. First, the environment can be modified so that it is less reactive with the test material. Second, the response of the material to the environment can be modified. Finally, the environment and the material can be separated so that no interaction can occur.

In this investigation, variations of these three basic approaches to preventing corrosion-fatigue were used to determine the most promising manner of alleviating the corrosion-fatigue problem in the offshore-drilling environment. In the first part of this paper, several modifications of the drilling environment were examined. These included the use of deaerated seawater, various drilling muds, and seawater mud with an elevated pH. In the second portion of this paper, methods of changing the material response to the environment by means of cathodic protection and the application of a sacrificial metal coating were examined. In the final section of the paper, results of the investigation are summarized and the more favorable methods discussed.

Experimental Program

The material used in this investigation was Type 135 drill-pipe steel with the typical composition given below:

Element	С	Mn	Р	S	Si	Mo
Percentage	0.45	1.61	0.018	0.030	0.22	0.17

This material was selected because it represented one of the higher strength production-line drill-pipe materials. The material was obtained in the form of standard 4-in, in outside diameter drill pipe that had been given the standard

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FIG. 1-Unnotched fatigue-specimen configuration.

heat treatment consisting of a quench from 1550 to 1600 F at a quenching rate of 35 ft/min followed by tempering at 1025 F. The average tensile properties based on triplicate tests were:

Ultimate strength, 159,200 psi

0.2 percent yield strength, 144,100 psi

Percentage elongation in 2-in. gage length, 12.5 percent.

All specimens were machined from the pipe section oriented parallel to the longitudinal axis of the pipe. The configuration of the unnotched fatigue specimens used in this program is shown in Fig. 1. The flat surfaces of the specimens were polished with 400-grit emery paper and kerosene. The edges of the specimens were polished on a wheel in steps using 240, 400, and 500-grit emery paper with kerosene. The resulting finish had a surface roughness of approximately 10μ in. rms, with all final polishing marks oriented parallel to the specimen axis.

All fatigue tests were conducted in a Krouse 10,000-lb capacity axial-load fatigue machine equipped with a mechanical speed reducer. This allowed the cyclic frequency of the test to be varied from 12 to 1200 cpm. For the corrosion-fatigue tests, a lucite container was fitted around the specimen and sealed at the bottom with silicon grease. The environment was allowed to flow into the bottom of the container and exit at the top so that a minimum of 2 in. of the environment covered the specimen test section at all times. All fatigue tests were conducted at a mean stress of 75 ksi.

Experimental Results

Control Test Results

Control tests were conducted on unnotched specimens in air and in seawater to provide baseline data against which to evaluate the effectiveness of the various protection methods. It has been demonstrated by Endo and Miyao [I]^a that, because the corrosion process is time-dependent, the fatigue

³ The italic numbers in brackets refer to the list of references appended to this paper.

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life obtained in a corrosive environment will be a function of the testing frequency. In order to assure that the frequency was slow enough to allow corrosion to be a factor, preliminary tests were conducted in seawater (see Appendix for composition) at frequencies varying from 12 to 1200 cpm. The results presented in Fig. 2 show only a slight difference between the results obtained at 12 and 60 cpm. The longer fatigue lives obtained at 1200 cpm, however, indicated the corrosion did not have sufficient time to show an effect due to the short duration of the test. On this basis, a frequency of 60 cpm was selected for use in all the corrosion-fatigue tests. The air-fatigue tests were conducted at a speed of 1200 cpm. The final results of the control tests in air and seawater are shown in Fig. 3.

Effect of Modified Environments

The effect of environment on the corrosion behavior of a metal is obvious; if the environment can be made more neutral, the corrosion is expected to be reduced. However, the exact nature of corrosion fatigue as it occurs in drilling fluids is not a simple process, but instead is a combination of the effects of several contaminates including oxygen, hydrogen sulfide, carbon dioxide, and various acids [2]. The following modifications of environment were examined to determine the effect of various factors that could be expected to influence the corrosiveness of the environment.

Deaerated Seawater—The role of oxygen in seawater often has been suspected of accelerating the corrosion-fatigue process [2]. To examine this possibility, fatigue tests were conducted using seawater deaerated with 1.6 g



FIG. 2—The effect of test frequency on the corrosion-fatigue lives of unnotched Type 135 steel.



FIG. 3-Axial load unnotched fatigue results.

of K-91 (calcium sulfite catalyzed with cobalt chloride) for every 5 gal of seawater. Analysis of the freshly deaerated seawater showed less than 0.02 ppm of oxygen as compared with the 5 to 6 ppm oxygen in the as-received seawater. Initial tests conducted in the open system showed no improvement of the fatigue properties. However, oxygen analysis of the seawater taken during the test indicated that some oxygen was being absorbed from the air, so a completely closed system, shown schematically in Fig. 4, was subsequently constructed. Checks of the oxygen level made during the tests in the closed system showed no change in the oxygen concentration from the initial value of less than 0.02 ppm. As shown in Fig. 5, no difference between the deareated and the as-received seawater test results was observed, thus indicating that reducing the oxygen level from 6 to 0.02 ppm does not affect the corrosion-fatigue process in seawater.

Drilling Muds—Fatigue tests employing either freshwater or seawater drilling mud (pH = 9 ± 0.2) were conducted in an open system in which the mud from the main reservoir was continuously circulated around the specimen. More detailed information on these muds is presented in the Appendix.

For the initial tests, a pH of 9 was selected since some results [3] indicate this may be an optimum value for the protective effect of colloids. Unnotched fatigue tests were conducted using freshwater mud with the pH of 9 and the fatigue results were found to be approximately equal to or slightly below the seawater-fatigue results. Similarly, tests conducted with the seawater mud, pH = 9, also were found to yield fatigue lives below those obtained at the same stress level in seawater, as shown in Fig. 6.





. It has been indicated [4] that improved corrosion-fatigue behavior may be obtained at high values of pH, so tests were conducted using the seawater mud with the pH adjusted to 11.6 to 12 by the addition of sodium bydroxide. Again, as shown in Fig. 6, the resulting fatigue lives were shorter than those observed for regular seawater tests.



FIG. 5-Fatigue behavior of unnotched specimens tested in deaerated seawater at 60 cpm.



FIG. 6 - Fatigue results in freshwater and seawater drilling muds.

Because neither of the drilling muds tested was found to have any advantage over seawater, an inhibitor known as Cortron DF-18, an oil-soluble amine-type inhibitor, was added to the seawater mud. This type of inhibitor is designed to produce a non-conducting film of organic polar compounds on the surface of the material. Prior to starting the tests, a mixture of 1 part inhibitor to 10 parts diesel oil was fed into the test chamber to cover the specimen. The chamber was then flushed with drilling mud and the test initiated. A one-to-one mixture of diesel oil and inhibitor was added to the drilling mud every 6 hours during the test. However, the results, shown in Fig. 6, showed no substantial improvement in the fatigue behavior in the inhibited mud over that observed in the regular mud.

Modifying Material Response to the Environment

Efforts to change the corrosive nature of the drilling environment had met with no success, so attention was turned to the possibility of modifying the material response to the environment. Several possible methods including cathodic protection and the application of sacrificial metal coatings were examined.

Cathodic Protection—The application of current to a material to eliminate corrosion has long been recognized as a potentially useful approach since corrosion is primarily electrochemical in nature. Tests were conducted by applying different applied currents to fatigue specimens being tested in the recirculating seawater system. The current was applied through platinum anodes submerged in the electrolyte and continuously monitored with a Calomel reference cell. These tests were referred to as "unagitated tests." A mechanical stirrer was then added to the system to produce increased velocities in the seawater. These tests are referred to as "agitated seawater tests."

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Specimens were tested in unagitated seawater and the results obtained are shown in Fig. 7. As shown, fatigue lives equivalent to those obtained in air could be produced at current densities from 40 to 100 mA/sq ft. During testing, it was observed that calcareous deposits formed on the specimen surface, the thickness and extent of these coatings increasing with increasing current density. This calcareous deposit consisted principally of calcium carbonate (CaCO₂) and magnesium hydroxide (Mg(OH)₂) and is believed to occur as a result of an increase in the pH of the electrolyte adjacent to the cathode [5]. The increase in pH presumably accompanies the flow of protective current.

Various factors govern the deposition of such coatings, and Humble [6] has studied the phenomenon in some detail. He concluded that the ratio of calcium carbonate to magnesium hydroxide in the coating is dependent on the applied current density, the composition of the water, and the amount of electrolyte turbulence at the cathode. This is in agreement with the results found by Nichols [7] for mild steel in seawater.

The effect of electrolyte velocity on the current density required for protection can be seen from the results plotted in Fig. 8. Again, the results indi-



FIG. 7—Cathodic-protection results on Type 135 steel, fatigue-tested at 60 cpm in unagitated seawater at a maximum stress level of 135,000 psi,

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FIG. 8— Cathodic-protection results on Type 135 steel, fatigue-tested at 60 cpm in agitated seawater at a maximum stress level of 135,000 psi,

cate that a rather substantial increase in lifetime results from the application of cathodic protection. However, under the condition of agitation, the current density required for protection has shifted from the 40 to 100 mA/sq ft range for the unagitated tests, to a value greater than 180 mA/sq ft for the agitated condition. The observed behavior is not readily explainable. It may be noted, however, that under agitated conditions no calcareous deposits became visible until a value of 200 mA/sq ft was reached. When the calcareous deposit formed at 200 mA/sq ft, it was of a different nature than the deposit formed in the unagitated tests. The important result is, however, that even under high degrees of agitation, improved fatigue behavior was achieved by cathodic protection.

Sacrificial Metal Coatings

Normally, protective coatings, such as paint, are designed to keep the corrodant from contacting the material. These coatings, however, have the inherent danger of becoming broken, thus allowing corrosion to proceed. For this reason, it was believed that a sacrifical metal coating that is anodic to the base metal would be more reliable, since even if rupture of the coating did occur, eathodic protection would still exist over the exposed area. In this

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manner, the inherent advantages of both the protective coating and cathodic protection could be employed.

As a means of selecting a potential sacrificial metal coating, a specimen of Type 135 steel was exposed without stress in seawater. A potential was established between the steel cathode and a platinum anode so that an applied current flowed between the platinum and the steel. As the applied current was varied, potential measurements were made between the steel and a saturated Calomel half cell at various current values. From measurements for both increasing and decreasing current, Fig. 9 was prepared. This figure shows the half-cell potential as a function of the current density at the steel for the data in the range that appears to be of interest from a fatigue standpoint. The actual measurements encompassed a range of current densities from -1800 to ± 600 mA sq ft. The vertical lines with solid circles at the end points represent specmens that were tested in fatigue under agitated cathodic-protection conditions at a constant current density. The solid circles represent the extremes in potential measured during the tests.

In Fig. 9, the range from 40 to 120 mA sq ft is cross-hatched and corresponds to the range where cathodic protection improved fatigue behavior. On the basis of (1) the curve of Fig. 9, (2) the current densities required for protection against corrosion fatigue, and (3) the potentials required to produce these current densities, the choice of metal coating for sacrificial protection is zinc and possibly alloys of zinc. Also indicated in Fig. 9, in the double-cross-hatched region, is the approximate range of the closed-circuit potential for the steel-zine couple.



FIG. 9-Potential versus current density relationship for Type 135 steel in seawater.

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Also shown in Fig. 9 is an arrow pointing to one of the curves at the current density value of 80 mA/sq ft. This indicates that hydrogen was evolved at this value. The important point is that, in these experiments, cathodic protection or protection by sacrificial zinc coatings occurs at current densities where atomic hydrogen is released at the cathode. Consequently, hydrogen damage is a potential problem.

Flame-Sprayed Zinc—As a first approach to applying the zine coating, a commercial zine flame-spraying procedure was used. Prior to the flame spraying, the specimen surface was grit-blasted to aid in bonding the zine. For this reason, it was first necessary to determine the effect of grit-blasting per se on the fatigue behavior of Type 135 steel.

Results of the tests on grit-blasted and grit-blasted plus flame-sprayed specimens are shown in Figs. 10 and 11, respectively. Figure 10 shows a marked decrease in the fatigue strength of the grit-blasted steel tested in air. A still further reduction in fatigue strength occurs for the grit-blasted specimens when tested in seawater. The reduction in strength noted in seawater for the grit-blasted specimens is not surprising since the surface damage produced by grit-blasting would be expected to be more susceptible to corrosion than unworked material. The reduction of the fatigue strength in air due to grit-blasting however, was not expected and indicates that the blasting treatment was very severe.

As seen in Fig. 11, the zinc coating does produce some increase in fatigue life in seawater over that of the grit-blasted material. The results show no increase in fatigue life over those obtained on bare specimens when tested in unagitated seawater, and only a slight improvement at low stresses when tested in agitated seawater. The improvement, however, was not as much as had been expected, probably due to the severe grit-blasting preparation used prior to coating.



FIG. 10-Effect of grit-blasting on the fatigue behavior of Type 135 steel.

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FIG. 11-- Fatigue behavior of zinc-coated (flame-sprayed) Type 135 steel tested in seawater.

Electroplated Zinc—Since the flame-spraying procedure did not substantially improve the fatigue behavior, the application of the zinc coating by electroplating was examined. Unnotched fatigue specimens were electroplated with zinc using the following procedure.

The specimen's were mounted in a plate rack built with non-conducting edge shields to reduce zinc thickness at the specimen edges and improve the thickness uniformity. The specimens were then cleaned and plated as follows:

1. Immersed in a 180 F chelating-type steel cleaner for a few seconds.

2. Anodically cleaned in the same cleaner with a current density of approximately 100 amp sq ft for 30 s.

3. Rinsed in tap water at 140 to 150 F.

4. Rinsed in tap water at 65 to 70 F.

5. Anodically treated in 10 percent (by weight) sulfuric acid solution at a current density of approximately 100 amp sq ft (3.5 v) for 10 s. A jumper contact was connected to each specimen before immersion in the acid solution to avoid chemical attack of the steel that might create hydrogen that could be absorbed by the steel.

6. Rinsed in an overflowing tap-water bath (65 to 70 F) while an anodic potential of 3.5 v was maintained to prevent chemical attack.

7. Plated with 4.0 \pm 0.5 mils of zinc in a high-efficiency zinc sulfate bath prepared with:

Zine sulfate (ZnSO ₁ ·7H ₂ O)	365.0 g/l
Aluminum sulfate (AISO ₁ -18H ₂ O)	53.3 g/l
Zine chloride (ZnCl ₂)	17.0 g l
Sodium sulfate (Na ₂ SO ₃).	75.0 g.1
Wetting agent (Tergitol 08)	15.0 ml/1
pH (adjusted with NaOH)	3.7 pH

The above bath was maintained at 85 ± 2 F during the plating operation. A mechanical stirrer was used to agitate the solution, and each specimen was plated for 70 min at a current density of 40 amp sq ft to deposit the desired thickness. One of the plated specimens was sectioned to determine the coating thickness, which was found to be 4 ± 0.5 mils. A plating thickness of 4 mils was selected because it could be expected to result in good adhesion and was commercially feasible.

The fatigue results obtained for the plated specimens in air and in seawater are shown in Fig. 12. From this figure, it can be seen that, while the electroplating results in severe degradation of the air fatigue results, a marked improvement in the corrosion-fatigue behavior in seawater is produced.

Since the coating may be broken during the pipe lifetime, several plated specimens were prepared with bare spots (hereafter referred to as "holidays"). This configuration is shown in Fig. 13. Holiday sizes ranging from $\frac{1}{5}$ to $\frac{3}{4}$ in.



FIG. 12 – Fatigue behavior of unplated and zinc-electroplated unnotched fatigue specimens of Type 135 steel tested in segwater and air.







FIG. 14—Fatigue behavior of zinc-electroplated unnotched fatigue specimens of Type 135 steel possessing various size holidays.

were tested in unagitated seawater to determine the maximum size defect tolerable in the zinc plating without a loss in fatigue strength. The results of these fatigue tests are shown in Fig. 14. If the results shown in Figs. 12 and 14 are compared, it can be seen that the presence of holidays up to $\frac{3}{4}$ in. still result in nearly complete protection at a maximum stress of 110 or 115 ksi. At the higher stress of 130 ksi maximum, some reduction in fatigue life does occur due to the holidays although the behavior is still superior to that of the unplated specimens.

Summary

The objective of this study was to examine a variety of approaches to the prevention of corrosion fatigue in drill pipe in the hope of determining the more promising procedures. For this reason, a variety of data was generated with the aim of determining the most promising direction for future efforts. On the basis of these results, the following trends are observed.

First, the use of certain modified environments to control corrosion fatigue does not appear to be a promising avenue of exploration. While only a few of many inhibitors and drilling muds now available were actually examined, no increase in fatigue life was observed for any of the methods examined. Second, a marked improvement in corrosion-fatigue life was noted with cathodic protection and with a sacrificial metal coating.

The results obtained with the sacrificial metal coating (zinc) indicate that this method of protection is very promising. While the results obtained for the flame-sprayed zinc coating were disappointing, they emphasize the importance of the method of application of the coating. On the other hand, the electroplated zinc coating was found to be quite effective in improving the corrosion-fatigue behavior. The tolerance of bare regions up to $\frac{3}{8}$ in. with no effect, and up to $\frac{3}{4}$ in. with little effect on the fatigue behavior indicates that the main benefit of the zinc coating lies in its electrochemical sacrificial nature.

The changes in the calcareous deposit produced by variations in the degree of agitation of the drilling fluid during cathodic protection do appear to have a bearing on the corrosion-fatigue behavior. In particular, the nature and effect of the deposit on possible hydrogen damage as related to the degree of agitation of the environment could prove to be extremely important, since the disadvantages produced by the hydrogen damage in the steel may outweigh the advantage obtained from the protection procedure. Further evaluation of this subject is needed to determine if hydrogen damage is a real problem in cathodic protection of drill pipe.

Conclusions

On the basis of the results obtained in this program, the following conclusions can be drawn:

1. The use of the following modified environments did not result in improvement in the corrosion-fatigue behavior over that obtained in seawater.

- (a) seawater deaerated with K-91
- (b) freshwater drilling mud, pH = 9
- (c) seawater drilling mud, pH = 9
- (d) seawater drilling mud, pH = 12

2. The use of cathodic protection can result in corrosion-fatigue lives in seawater equivalent to those obtained in air.

3. The use of an electroplated zinc coating results in an improvement in corrosion-fatigue life, but a decrease in the air-fatigue life.

4. Bare regions of steel up to $\frac{3}{4}$ in. can be tolerated in the zinc-coated specimens without appreciable loss of protection.

5. The use of a flame-sprayed zinc coating did not result in a general improvement in corrosion-fatigue behavior. This is believed to be partially the effect of the severe grit-blasting the material received prior to flame-spraying.

In future work the effectiveness of new methods such as plastic coatings, special surface treatments, new corrosion-resistant materials, new inhibitors, etc., should be examined in an effort to produce a more suitable corrosion-resistant drill pipe.

Acknowledgments

The research program upon which this paper is based was funded by Brown & Root Company under a subcontract from the National Science Foundation. Sincere thanks are given to Brown & Root Company for their cooperation in the conduction of this program.

APPENDIX

Composition and Preparation of the Drilling Muds Employed

The drilling muds employed were supplied by the Baroid Division of the National Lead Company.

Composition

Composition of 5 gal of each of the two drilling muds is given in Table A1.

Freshwater N	1ud		Seawater M	ud	
Deionized Water	17 500	ml	Seawater"	17 500	m
Aquagel .	1 250	g	Aquagel	2 000	g
Q-Broxin.		g	Zeogel	350	g
Caustic Soda	7.5	5 g	Q-Broxin	62.5	g
			Caustic Soda	12.1	g

TABLE A1-Composition of drilling muds, per 5 gal.

" 42 g "sea-salt" per liter of deionized water.

Preparation

Freshwater Mud. The freshwater mud was prepared in separate 5-gal batches. Each batch was prepared by adding 1250 g of Aquagel to 17,500 ml of deionized water and stirring at high speed with a dispersator for 1 h. The resulting slurry was treated with 45 g of Q-Brexin and 5 g of caustic soda and stirred an additional hour. The 5-gailon batches were biended and heated to 180 F while stirring for 3 h. After the mud had cooled, the pH and viscosity were adjusted with the addition of 35 g of Q-Broxin and 2.5 g of caustic soda. The mud was then stirred for 3 h at high speed with a dispersator.

Seawater Mad. The seawater mud was prepared in separate 5-gal batches. Each 5-gal batch was prepared by adding 2000 g of Aquagel and 350 g of Zeogel to 17,500 ml of seawater and stirring at high speed with a dispersator for 1 hour. The resulting slurry was treated with 62.5 g of Q-Broxin and 6.1 g of caustic soda and stirred for an additional hour. The two batches of mud were blended and heated to 180 F while stirring for 3 h. After the mud had cooled, the pH was adjusted by the addition of 6 g of caustic soda. The mud was then stirred for 3 h at high speed with a dispersator.

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APPENDIX C

Mechanics of Long Drill Strings

MECHANICS OF LONG DRILL STRINGS

P. H. Francis P. A. Cox

FINAL REPORT SwRI Project No. 02-2807

Prepared for

Deep Sea Drilling Project Scripps Institution of Oceanography La Jolla, California

August 17, 1970



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Approved:

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NOMENCLATURE

Roman

A	cross-sectional area of pipe
С	damping force per unit length per unit velocity distance from neutral axis to outer fiber (radius)
<i>c</i> ₀	defined by Eq. (2.14)
<i>c</i> ₁	defined by Eq. (2.9)
С	concentrated damping force per unit velocity
C_D	coefficient of drag
d	diameter
D	distributed damping force per unit length per unit velocity cumulative damage
Ε	elastic modulus
F	defined by Eq. (2.8) fractional fatigue damage
g	gravitational acceleration
h	height of guide shoe
Н	swell height, double amplitude
Ι	second moment of area of drill pipe section
k	added mass coefficient
Κ	corrosion factor
l	length of drill string
L	fatigue life, hours
М	concentrated end mass bending moment
n	number of reversed bending cycles number of drag (damping) disks

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Ν	defined by Eq. (2.7) number of rubbers number of cycles to failure
r	radius
R	radius
S	rubber spacing
t	time
Т	swell period tension in pipe
T_{1}, T_{2}	defined by Eqs. (2.15) and (2.16)
и	absolute string (extensional) displacement
v	relative string (extensional) displacement
V	response velocity shear force
W	displacement of ship
x	axial coordinate
ż	drilling rate
X	response amplitude
У	bending displacement
Greek	
α	defined by Eq. (2.17) defined by Eq. (3.4)
α1	defined by Eq. (2.10)
β	defined by Eq. (2.11) angle through which drill string bends
γ	defined through Eq. (2.12)
Г	Gamma function
δ	Dirac delta function
θ	roll/pitch angular deviation

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λ	swell wave length
μ	distributed mass density of string
ν	kinematic viscosity
ρ	mass density
σ	stress
φ	angle through which picalo bends
Φ	power spectral density
ω	frequency
ω_1	defined by Eq. (2.13)
Ω	ship response frequency

Subscripts

a	added
d	design
e,eq	equivalent
f	fluid
т	maximum
р	pipe
r	rubber
rms	root mean square
v	vessel

SECTION 1. INTRODUCTION

This report contains the final results of a study concerned with determining the dynamic stresses and fatigue life of the drill string system of the Glomar Challenger, Deep Sea Drilling Project. The Challenger as now in use operates with a guide sheet of about 51.7 ft in height, which is installed just below the drilling platform, and which serves the purpose of limiting the bending stresses in the pipe as the ship rolls and pitches in response to sea motions. The guide shoe structure has thus far performed quite satisfactorily in regard to inhibiting excessive bending stresses in the pipe.

However, due to its physical size and "permanent" installation, the guide shoe has presented some rather difficult logistic problems in actual drilling operations. It is not possible to deploy or retract the drill string through the guide shoe with the re-entry tools and equipment at the lower end of the string, and it will frequently be necessary to keel-haul such equipment at drill sites where multiple core bit runs are required. This process of keel-hauling is both time-consuming and dangerous. Hence, it would be highly desirable to find an attractive solution which would make this phase of the operation unnecessary. This objective became the motivation of this program. It was desired to define as accurately as possible the loading environment on the drill string, the predicted fatigue life of the string under operating conditions, and to interpret these predictions in terms of alternative concepts for limiting the bending in the string.

The work of this program was based upon two reports submitted to Global Marine, Incorporated, by Dr. T. Vreeland, Jr.(1.1,1.2), consultant to Global Marine. These reports dealt, respectively, with the Heave Dynamics of the drill string and with the effects of roll and pitch ship motions on the drill string. These reports, and the results therein contained, served as a basis for design assurance of the drill string/guide shoe concept now being employed on the Challenger operations.

The first objective of the present work was to review these reports and to see if it was possible to remove any unusual conservatism in the stress and life predictions. Then, computer programs were written to solve the resulting equations, and the computed results were assessed in terms of alternate concepts to the guide shoe configuration. The overall analysis procedures set forth in the reports by Vreeland were found to be quite thorough and satisfactory. The present report, in fact, adheres quite closely to his presentation, and the mathematical notation is very nearly identical to that employed by Vreeland. There were, however, a number of departures made in the theoretical work, and, hence, the final stress and life prediction equations differ rather substantially in some cases from those presented by Vreeland. As a result of these differences in the present analysis vis-a-vis Dr. Vreeland's, the present results differ somewhat from his. In general, however, the differences between the two sets of results are not substantial and at any rate do not lead to contradictions with any of Vreeland's overall study conclusions.

The analytical formulations of this study are presented in Section 2, Heave Dynamics, and Section 3, Roll and Pitch Effects. The mathematical calculations therein contained are given in rather complete detail. In order to facilitate the reading of these sections, much of the detail derivations and ancillary information are relegated to smaller print and are presented as they occur. Section 4 presents the computed results for the Heave Dynamics, and for roll and pitch effects with both the guide shoe and picalo systems. These results are summarized in Section 5 and, based upon these summaries, a number of recommendations are made concerning possible alternate concepts to the present guide shoe system used on the Glomar Challenger. Section 6 contains a complete computer program documentation. Also included in this section are sample runs which may be used in checking out the numerical program on other computer systems. Appendix A contains a detailed derivation of the modal solution to the Heave Dynamics problem of the drill string. This solution is rather lengthy and the resulting equations differ somewhat in form from those presented by Vreeland(1.1)

SECTION 2: HEAVE DYNAMICS

In this section, there is presented an analysis of the dynamic response of the drill string to heave excitation. The dynamic stresses and displacements are computed both for deterministic and random sea states, and the fatigue life of the drill string is estimated on the basis of these stress computations.

The analysis is based on a model of the drill string assembly as a uniform elastic rod with a concentrated mass at the lower end representing the drill collars, drill, drag disks, and landing base. The string is excited harmonically at the upper end. Damping is included to account for the dissipation of the rod, rubbers, drag disks, and landing base. The modal solution of the differential equation of motion is deferred to Appendix A so as not to detract from the orderly presentation of results given in this section. *Other derivations, all of which are short, are set in small print as encountered.*

2.1 Dynamic Response of Drill String

The drill string is shown schematically in Figure 2.1.1. The string is excited at the top by a time-dependent displacement W(t). The absolute displacement of a segment of the string is u(x,t), where x is the space coordinate measured from the top of the string. The rod is considered to be uniform along its length in modulus, cross-sectional area, density, and damping capacity. With these assumptions, the response is given by solutions of the differential equation.

 $\rho A = \frac{\partial^2 u}{\partial t^2} = AE \frac{\partial^2 u}{\partial x^2} - c \frac{\partial u}{\partial t}$



(2.1) Figure 2.1.1–Schematic of Drill String

In this equation, ρ , A, E, and c are, respectively, the mass density, cross-sectional area, modulus, and damping force of the string (see Nomenclature). To account for both distributed and concentrated mass and damping, the following quantities are defined:

$$\rho A = \mu A + M\delta(x - \ell) \tag{2.2a}$$

$$c = D + C\delta(x - \ell) \tag{2...}$$

where $\delta(x - \ell)$ is the Dirac delta function. The boundary conditions which must be imposed on solutions of Eq. (2.1) are:

$$u(0,t) = W(t)$$
 (2.3a)

$$\frac{\partial u}{\partial x}\Big|_{x=0} = -\frac{M}{AE} \frac{\partial^2 u}{\partial t^2}\Big|_{x=0}$$
(2.3b)

The first of these is to insure the prescribed displacement at the top of the string, while the second balances the internal and inertia forces at the end of the string, to which the concentrated mass is affixed.

The problem just posed has been solved by a modal analysis for the time-dependent and maximum displacement, and stress. The analysis uses only the first vibration mode of the string.

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This approximation is acceptable for a long string under low-frequency excitation, as in the case of the drill string. The derivation of the response equations given below is presented in Appendix A. Reference should be made to the damping calculations later in this section and to Table 2.4.1 for correct interpretations of the constants in the following equations.

$$\sigma_m(x) = \left(\frac{ENc_1\alpha_1}{\varrho}\right) \cos\left(\frac{\alpha_1 x}{\varrho}\right)$$
(2.4)

$$v_m(x) = Nc_1 \sin\left(\frac{\alpha_1 x}{\varrho}\right)$$
 (2.5)

$$u_m(\ell) = \left\{ \left[W_0 + \frac{N}{F} (T_1 \cos \alpha - T_2 \sin \alpha) \right]^2 + \frac{N^2}{F^2} (T_1 \sin \alpha + T_2 \cos \alpha)^2 \right\}^{1/2}$$
(2.6)

where:

$$W(t) = W_0 \sin \Omega t$$

$$N = F \left\{ \left[\frac{M\omega_1^2 c_1^2}{\beta} \left(\frac{1}{2} + \frac{1}{4\alpha_1} \sin 2\alpha_1 \right) - \frac{\Omega^2}{\beta} \right]^2 + C^2 \Omega^2 c_1^4 \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_1} \sin 2\alpha_1 \right) + \sin^2 \alpha_1 \right]^2 \right\}^{-1/2}$$
(2.7)

$$F^{2} = W_{0}^{2}c_{1}^{2} \left\{ \frac{M^{2}\Omega^{4}}{\beta^{2}} \left[\frac{1}{\alpha_{1}} \left(1 - \cos \alpha_{1} \right) + \beta \sin \alpha_{1} \right]^{2} + C^{2}\Omega^{2} \left[\frac{\gamma}{\alpha_{1}} \left(1 - \cos \alpha_{1} \right) + \sin \alpha_{1} \right]^{2} \right\}$$
(2.8)

$$c_{1} = \left[M \left(\frac{1}{2} - \frac{1}{4\alpha_{1}} \sin 2\alpha_{1} + \beta \sin^{2} \alpha_{1} \right) \right]^{-1/2}$$
(2.9)

$$ctn \,\alpha_1 = \beta \alpha_1 \tag{2.10}$$

$$\beta = \frac{M}{\mu A \varrho} \tag{2.11}$$

$$\gamma = \frac{D\varrho}{C} \tag{2.12}$$

$$\omega_1 = \frac{\alpha_1 c_0}{\varrho} \tag{2.13}$$

$$c_0^2 = \frac{E}{\mu}$$
 (2.14)

$$T_1 = \frac{MW_0 c_1^2 \Omega^2}{\beta} \left[\frac{1}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \beta \sin \alpha_1 \right] \sin \alpha_1$$
(2.15)

$$T_2 = CW_0 c_1^2 \Omega \left[\frac{\gamma}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \sin \alpha_1 \right] \sin \alpha_1$$
(2.16)

$$\alpha = \operatorname{Tan}^{-1} \left\{ \frac{Cc_1^2 \Omega \beta \left\{ (\gamma/4) \left[2 - (1/\alpha_1) \sin 2\alpha_1 \right] + \sin^2 \alpha_1 \right\}}{M \omega_1^2 c_1^2 \left[(1/2) + (1/4\alpha_1) \sin 2\alpha_1 \right] - \Omega^2} \right\}$$
(2.17)

These equations enable one to determine the dynamic stress and displacement for a harmonic sea state for any set of drill string parameters.

2.2 Ship Response to Swells

Swells of height H (double amplitude) and frequency Ω cause heave motions of the vessel which can be expressed as

$$W(t) = H_{\nu}(\Omega) H \sin \Omega t \tag{2.18}$$

where $H_{\nu}(\Omega)$ is the transfer function for the vessel. Model tests for vessel response to 45° quartering swells (conservative design practice) have shown that a good approximation to the transfer function for the vessel is

$$H_{\nu}(\Omega) = 0.08 (T-2), \ 2 < T < 14$$
 (2.19)

where $T = 2\pi/\Omega$ is the swell period in seconds.

It is necessary now to derive an expression for the maximum swell height for a given swell period. There are two ways to proceed. Strictly speaking, the maximum swell height is to be found on the basis of a trochoidal wave form. Such an analysis gives the following relationship for the maximum stable swell height:

$$H_{\rm max} = 1.30T^2 \tag{2.20}$$

This relationship, however, probably represents overly-conservative design practice, since drilling operations are not likely to be done in seas of trochoidal form ("rough" seas) with wave heights high enough to be damaging to the drill string.

Equation (2.20) can be derived as follows. Lamb(2.1), p. 417, shows the finite-amplitude surface wave velocity to be

$$= \sqrt{\frac{g}{k} (1 + k^2 H_{\max}^2)}$$
$$= \sqrt{\frac{\lambda g}{2\pi} \left[1 + \left(\frac{2\pi H_{\max}}{\lambda}\right)^2 \right]} , \text{ since } k = \frac{2\pi}{\lambda}$$

In these expressions, c is the wave velocity, g the gravitational acceleration constant, λ the wave length, and H_{\max} the maximum wave amplitude. Now, since $\lambda = cT$:

$$\lambda = \frac{gT^2}{2\pi} \left[1 + \left(\frac{2\pi H_{\max}}{\lambda} \right)^2 \right]$$

Finally, the ratio of wave amplitude to wave length, H_{max}/λ , is to be found in the footnote on p. 418 of Lamb (due to a printing error, the decimal is missing):

$$\frac{H_{\max}}{\lambda} = 0.142$$

for trochoidal wave profiles. Thus, on substituting,

$$H_{\max} = 0.142 \frac{gT^2}{2\pi} \left\{ 1 + [2\pi(0.142)]^2 \right\}$$
$$H_{\max} = 1.30T^2$$

A somewhat more realistic relationship may be found by assuming a sinusoidal wave profile. For the sinusoidal wave, the maximum stable swell height is found to be

$$H_{\rm max} = 0.817T^2 \tag{2.21}$$

This is the relationship which has been used as the basis of the heave dynamics loading environment for purposes of this report. It turns out, however, that the maximum stable sinusoidal swell is unconservative as a design criterion, for fractional values of the coefficient 0.817 lead to higher dynamic stresses and shorter fatigue lives. This point is discussed in more detail in Section 4. Computed Results.

Equation (2.21) can be found very simply by postulating that, at the crest of the maximum stable swell, the pressure is zero, so that the acceleration and gravitational forces exactly annihilate each other:

 $\ddot{H} = g$

thus,

$$\Omega^{*}H_{\max} = g$$

$$H_{\max} = \frac{g}{\Omega^2} = \frac{gT^2}{(2\pi)^2} = 0.817T^2$$
 (ft)

Combining Eqs. (2.18) and (2.21), there results the following expression for the maximum vessel response as a function of sea period:

$$W_{\rm max} = 0.0653 \, (T-2)T^2 \tag{2.22}$$

The question arises as to whether or not one is justified in ignoring the force supplied by the drill string when computing the ship response to swells, as has been done in this report. A simple calculation will show that the ship displacement due to the drill string force is insignificant with respect to the ship displacement in swells, thus justifying the assumption. The dynamic force imparted to the vessel by the string is $F_s = \sigma A = 5.80\sigma$, where σ is the dynamic stress and a 19.5 lb/ft string has been assumed. This force creates a vertical displacement of the ship, which can be estimated by

$$W_{\rm s} = \frac{F_{\rm s}}{(M+M_{\rm a})\ \Omega^2}$$

where M and M_a are, respectively, the actual mass and the added mass of the ship, and Ω is the ship frequency:

$$M = \frac{13.4 \times 10^6}{32.2} = 0.416 \times 10^6$$
$$M_a = \pi \rho b^2 Q = \pi \left(\frac{64.4}{32.2}\right) (65)^2 (400) = 10.6 \times 10^6$$
$$\Omega = \frac{2\pi}{T} = 1.257 \quad \text{(for } T = 5 \text{ sec)}$$

Thus, $W_s = 3.34 \times 10^{-7} \sigma$. The vertical displacement as calculated without consideration of the drill string, W_a is

$$W_a = 0.08(T-2)(0.817T^2) = 4.9$$
 ft = 58.9 inches

Thus, comparing W_s with W_a :

$$\frac{W_s}{W_a} = 5.67 \times 10^{-9} \sigma$$

which is insignificant for any realistic value of σ . The same conclusion holds for other assumptions regarding M_a and T than those assumed here.

2.3 Ship and String Response to a Random Sea

In this section, expressions for the root mean square (rms) values of the maximum stress in the drill string, period of the stressing, and displacement of the bottom of the string, are presented. The following customary assumptions are used:

- (1) The sea motion is weakly stationary
- (2) The sea motion is Gaussian
- (3) The ship transfer function is linear.

Under these three assumptions, the excitation of the drill string is Gaussian and weakly stationary. The displacement, velocity, and acceleration power spectral densities of the vessel response are defined in terms of the power spectral density of the sea, Φ_{SS} , as

$$\Phi_{WW}(\omega) = \Phi_{SS}(\omega) \left| H_{\nu}(\omega) \right|^2 \tag{2.23}$$

$$\Phi_{\dot{W}\dot{W}}(\omega) = \omega^2 \Phi_{SS}(\omega) |H_{\nu}(\omega)|^2$$
(2.24)

$$\Phi_{\widetilde{W}\widetilde{W}}(\omega) = \omega^4 \Phi_{SS}(\omega) |H_{\nu}(\omega)|^2, \qquad (2.25)$$

respectively. In these expressions, ω is the excitation frequency in radians/second, Φ_{SS} is the power spectral density of the sea wave height, and $H_{\nu}(\omega)$ is the ship transfer function, Eq. (2.19).

The power spectral density of the drill string response is obtained by finding $\Phi_{\eta\eta\dot{w}}$ and $\Phi_{\eta\eta\dot{w}}$. respectively, the power spectral densities resulting from the \dot{W} and \ddot{W} excitations. These functions are:

$$\Phi_{\eta\eta\dot{w}} = c_1^2 C^2 \left[\frac{\gamma}{\alpha_1} (1 - \cos \alpha_1) + \sin \alpha_1 \right]^2 \left[H_S(\omega) \right]_1^2 \Phi_{\dot{w}\dot{w}}(\omega)$$
(2.26)

$$\Phi_{\eta\eta}\ddot{w} = \frac{c_1^2 M^2}{\beta^2} \left[\frac{1}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \beta \sin \alpha_1 \right]^2 \left[H_S(\omega) \right]^2 \Phi \ddot{w} \ddot{w}(\omega)$$
(2.27)

where

$$|H_{S}(\omega)|^{2} = \frac{N^{2}}{F^{2}}$$
$$= \left\{ \left[\frac{M\omega_{1}^{2}c_{1}^{2}}{\beta} \left(\frac{1}{2} + \frac{1}{4\alpha_{1}}\sin 2\alpha_{1} \right) - \frac{\omega^{2}}{\beta} \right]^{2} + C^{2}\omega^{2}c_{1}^{4} \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_{1}}\sin 2\alpha_{1} \right) + \sin^{2}\alpha_{1} \right]^{2} \right\}^{-1} \quad (2.28)$$

Note that $\Phi_{\eta\eta\dot{w}}$ and $\Phi_{\eta\eta\dot{w}}$ are components of the response power spectral density and are not themselves the velocity and acceleration power spectral densities.

With these preliminaries, the rms values of top end stress, lower end relative displacement, and frequency are given as follows:

$$\sigma_{\rm rms}(0) = E\eta_{\rm rms}c_1\alpha_1/\varrho \tag{2.29}$$

$$\gamma_{\rm rms}(\ell) = \eta_{\rm rms} c_1 \sin \alpha_1 \tag{2.30}$$

$$\omega_{\rm rm\,s} = \left[\int_{0}^{\infty} \omega^2 \left(\Phi_{\eta\eta}\dot{w} + \Phi_{\eta\eta}\ddot{w}\right) d\omega / \eta_{\rm rm\,s}^2\right]^{1/2}$$
(2.31)

where

$$\eta_{\rm rm\,s}^2 = \int_0^\infty \left(\Phi_{\eta\eta\,\dot{w}} + \Phi_{\eta\eta\,\ddot{w}} \right) \,\mathrm{d}\omega \tag{2.32}$$

These quantities are all determined completely from the known dynamic drill string response and ship transfer function, once the sea power spectral density has been specified.

The power spectral density of the sea state, Φ_{SS} , may be described either by data taken directly from a spectral analysis, or by an approximating function which characterizes the main features of the sea states of interest. In this report, the second alternative has been chosen. Ochi(2.2) has presented power spectral density data typical of those encountered in moderate to heavy seas. The energy content of these spectra lies principally in the band $0.2 < \omega < 1.0$. A reasonable approximation to these data is the function

$$\Phi_{SS}(\omega) = A \left\{ 1. - \cos\left[\frac{5\pi}{2}(\omega - 0.2)\right] \right\}^2$$
(2.33)

The constant A corresponds to the maximum amplitude, in ft^2/rad of the power spectral density and occurs at $\omega = 0.6$. The integrated spectral energy is

$$\int_{0.2}^{1.0} \Phi_{SS}(\omega) \, \mathrm{d}\omega = \frac{64}{5}$$
(2.34)

so that A is 5/6 of the integrated spectral energy in units of ft^2 sec.

2.4 Damping Calculations

Damping is introduced into the drill string system through the drag and added mass effects of the drill string, rubbers, drag disks, and end mass. Each of these effects has an important influence on the dynamic response of the string.

The differential equation of motion of the system, as given before [Eq. (2.1)], is

$$\mu A \frac{\partial^2 u}{\partial t^2} + M\delta(x-\ell) \frac{\partial^2 u}{\partial t^2} = AE \frac{\partial^2 u}{\partial x^2} - D \frac{\partial u}{\partial t} - C\delta(x-\ell) \frac{\partial u}{\partial t}$$
(2.35)

The drag coefficient D is composed of contributions D_1 and D_2 from the drill pipe and from the rubbers. In addition, the rubbers are considered to contribute an added mass effect to the first term in the differential equation above. The drag disks at the lower end of the string give rise both to the drag coefficient C and an added mass effect to the second term in the differential equation.

Consider first the distributed damping coefficient $D: D = D_1 + D_2$, where D_1 and D_2 are the velocity-damping coefficients of the drill string and the rubbers, respectively. The drill string is assumed to oscillate harmonically, so that at any point the displacement X and velocity V are, respectively,

$$X(t) = X_o \sin \omega t$$
$$V(t) = X_o \omega \cos \omega t$$

It will be shown below that D_1 and D_2 are expressible as

$$D_1 = \sqrt{2\nu\omega} \pi r \rho_f \tag{2.36}$$

$$D_2 = \frac{\rho_f}{2} n_2 C_D \left(\frac{\pi d_r^2}{4}\right) X_0 \omega \tag{2.37}$$

In these expressions, ν is the kinematic viscosity, r the radius of the drill pipe, ρ_f the mass density of the sea water, n_2 the number of rubbers per unit length of pipe, C_D the coefficient of drag of a flat disk, and d_r the diameter of the rubber. From potential flow theory, the coefficient of drag C_D for a thin disk is $C_D = 1.12$ for Reynold's numbers $N_R > 10^2$, as in the present case.

To derive Eqs. (2.36) and (2.37), reference is made first to Lamb(2.1), who shows that the force/unit area f(t) that a plate oscillating harmonically about the x-y plane imparts to a semi-infinite pipe $y \ge 0$ is:

$$f(t) = \rho_f \sqrt{\nu \omega} X_0 \omega \cos \left(\omega t + \frac{\pi}{4}\right)$$

The work/unit cycle done by the plate on the pipe is

$$\operatorname{vork} = \int_{\operatorname{cycle}} f(t) \, \mathrm{dx} = \int_{\operatorname{cycle}} f(t) \, X_0 \omega \cos \omega t \, \mathrm{dt}$$
$$= \int_0^{2\pi/\omega} \rho_f \, \sqrt{\nu\omega} \, (X_0 \omega)^2 \cos \left(\omega t + \frac{\pi}{4}\right) \cos \omega t \, \mathrm{dt} = \rho_f \, \sqrt{\nu\omega} \, (X_0 \omega)^2 \, \frac{\sqrt{2} \, \pi}{2\omega}$$

This work/unit cycle is to be identical with an "equivalent" in-phase damping:

$$\rho_f \sqrt{\nu\omega} (X_0 \omega)^2 \frac{\sqrt{2\pi}}{2\omega} = \int_{\text{cycle}} D_e X_0 \omega \cos \omega t \, dx = D_e (X_0 \omega)^2 \int_0^{2\pi/\omega} \cos^2 \omega t \, dt = \frac{\pi D_e (X_0 \omega)^2}{\omega}$$

Thus, $D_e = \sqrt{2}/2 \rho_f \sqrt{\nu \omega}$. The damping coefficient per unit length, D_1 , then is found by multiplying the damping coefficient per unit area, D_e , by the circumference of the drill pipe, $2\pi r$. This gives the result

$$D_1 = \sqrt{2\nu\omega} \pi r \rho_f$$

stated as Eq. (2.36).

The rubbers are assumed distributed uniformly along the drill string length, and their effective damping coefficient, D_2 , may be computed in the following manner. The damping force F_2 is customarily taken as proportional to the square of the velocity:

$$F_{2} = C_{D} \frac{1}{2} \rho_{f} \left(\frac{\pi d_{r}^{2}}{4} \right) V \left| V \right| = C_{D} \frac{1}{2} \rho_{f} \left(\frac{\pi d_{r}^{2}}{4} \right) \left(X_{0} \omega \right)^{2} \cos \omega t \left| \cos \omega t \right|$$
$$\approx C_{D} \frac{1}{2} \rho_{f} \left(\frac{\pi d_{r}^{2}}{4} \right) \left(X_{0} \omega \right)^{2} \cos \omega t$$

The indicated approximation is justified by the close similarity of the functions $\cos \theta |\cos \theta|$ and $\cos \theta$. For N rubbers, the total damping force is NF_2 , and the damping constant per unit length is, therefore,

$$D_2 = \left(\frac{N}{\varrho}\right) \frac{F_2}{V} = n_2 C_D \frac{1}{2} \rho_f \left(\frac{\pi d_r^2}{4}\right) X_0 \omega$$

where $n_2 = N/\ell$

The rubbers also contribute an added mass effect because of the mass of water they must displace during oscillation. This added mass effect shows up as an added term to the
coefficient μA in the leading term of the differential equation of motion of the drill string. The acceleration force for one rubber is

$$F_3 = k\rho_f B \frac{\partial^2 u}{\partial t^2} = k\rho_f \left(\frac{\pi d_r^3}{6}\right) \frac{\partial^2 u}{\partial t^2}$$

where the volume of the added mass effect is assumed to be that of a sphere having the same diameter as the rubber. The added mass coefficient k is taken to be $2/\pi$, and, hence, the added mass of the rubbers, $M_a \partial^2 u / \partial t^2$, is

$$M_a \frac{\partial^2 u}{\partial t^2} = n_2 \rho_f \frac{d_r^3}{3} \frac{\partial^2 u}{\partial t^2}$$
(2.38)

The added mass coefficient k can be derived by considering the kinetic energy of the system

$$FV = \frac{\mathrm{d}T}{\mathrm{d}t} = M_a V \frac{\mathrm{d}V}{\mathrm{d}t}$$

where F, V, and T are, respectively, the force, velocity, and kinetic energy of the added mass. The kinetic energy $T = (4/3)(\rho_f r^3 V^2)$ from Milne-Thompson^(2.3). Thus,

$$M_a = \frac{1}{V} \frac{dT/dt}{dV/dt} = \frac{1}{V} \frac{dT}{dV} = 2\rho_f \cdot \frac{4}{3} r^3 = \frac{2}{\pi} \cdot (\text{mass of sphere})$$

In terms of the earlier notation,

$$M_a = \frac{2}{\pi}$$
 (mass of sphere) = k (mass of sphere)

and, hence,

$$k=\frac{2}{\pi}=0.637$$

This, incidentally, corresponds with the value stated in the paper by Iversen and $Balent^{(2.4)}$.

The drag disks at the lower end of the drill string also contribute damping and added mass effects. It is assumed that if more than a single drag disk is placed at the end of the string, they are separated enough so that their effects do not interact, yet they are close enough together so that they can be considered "lumped" with respect to the string length.

The coefficient of damping C is computed exactly as before for the damping coefficient D_2 of the rubbers. The result is

$$C = nC_D \frac{1}{2} \rho_f \left(\frac{\pi d_d^2}{4}\right) X_0 \omega$$
(2.39)

where n is the number of drag disks and d_d is the diameter (actual or effective) of the drag disks.

The added mass term, again, is computed just as was done for the rubbers. The added mass term is

$$M_{a}\delta(x-\ell)\frac{\partial^{2}u}{\partial t^{2}} = n\rho_{f}\frac{d_{d}^{3}}{3}\delta(x-\ell)\frac{\partial^{2}u}{\partial t^{2}}$$
(2.40)

From these results, the interpretations listed in Table 2.4.1 are to be given to the coefficients in the original differential equation of motion, Eq. (2.35).

These are the values of the coefficients for the modal analysis solution, Eqs. (2.4) to (2.17).

It should be noted that the damping coefficients depend both upon frequency ω and amplitude X_o . Therefore, an iterative solution was used in the response calculations, wherein the damping for a particular frequency was first calculated using an assumed amplitude, the amplitude was calculated using the response analysis presented in this section, and the damping constant recalculated using the new value for amplitude. This process was found to converge

TABLE 2.4.1 EXPRESSIONS FOR COEFFICIENTS IN EQ. (1.35)

Coeff. in Eq. (1.35)	True Expression
μА	$\mu A + n_2 \rho_f \frac{d_r^3}{3}$
М	$M + n\rho_f \frac{d_d^3}{3}$
AE	AE
D	$\sqrt{2\nu\omega}\pi r\rho_f + 1.12n_2\frac{\rho_f}{2}\left(\frac{\pi d_r^2}{4}\right)X_0\omega$
С	$1.12n\frac{\rho_f}{2}\left(\frac{\pi d_d^2}{4}\right)X_0\omega$

rapidly. The effect of variation in amplitude along the string on the distributed damping (due to the amplitude dependence of damping) was taken into account by assuming a mean amplitude of $u(\mathfrak{A})/\sqrt{2}$ for the case of swells, and $v(\mathfrak{A})/\sqrt{2}$ for the case of the random sea.

2.5 Fatigue Calculations

The drill pipe employed on the Glomar Challenger drilling operations is Grade S-135 Drill Pipe. having average yield and ultimate strength values of about 145.0 and 159.3 ksi, respectively*. The fatigue properties of this type of material were discussed in an article by Rollins(2.5), who summarized much of the recent fatigue data generated for S-135 drill pipe material. The data presented in the article by Rollins are used as a basis for describing, quantitatively, the fatigue behavior of the drill string for purposes of this report.

Rollins notes that the endurance limit in reversed bending for S-135 drill pipe in air is about 80 ksi, whereas there is no apparent endurance limit for specimens tested in sea water. For purposes of comparison, however, at 10^6 cycles, the reversed bending fatigue strengths are about 88 ksi in air and about 29 ksi in sea water, a reduction of approximately 2/3.

A conservative approximation to the data given by Rollins can be made for zero mean stress fatigue loading, assuming linearity between stress σ and the logarithm of N, the number of cycles to failure. The expression used in this report is

$$N(\sigma) = \left(\frac{11.023 \times 10^5}{\sigma}\right)^{3.82} \tag{2.41}$$

To account for the fact that a mean stress does exist in the drill pipe, use is made of the well-known Goodman equation to define an "equivalent" stress level σ_{eq} having no mean stress; see Madayag(2.6):

$$\sigma_{eq} = \frac{\sigma}{K(1 - \sigma_s/\sigma_u)}$$
(2.42)

^{*}These values were averages of nineteen specimens, whose strengths ranged from 136.6 to 153.6 ksi in yield and from 153.7 ksi to 166.4 in ultimate.

In this expression, σ is the alternating stress amplitude, σ_s the mean (static) stress in the pipe, σ_u the ultimate tensile strength, and K an empirical factor to account for the corrosive nature of the sea water environment. For all calculations presented in this report, a factor K = 0.9 was used. In view of Eq. (2.42), Eq. (2.41) becomes

$$N(\sigma) = \left(\frac{11.023 \times 10^5}{\sigma_{eq}}\right)^{3.82}$$
(2.43)

If the ship operates in a sea having swells of essentially constant amplitude, the life in hours, L, may conveniently be calculated by means of the formula:

$$L = \frac{2\pi N(\sigma)}{3600\omega_1} \tag{2.44}$$

where $\omega_{\tilde{t}}$ is given by Eq. (2.13).

For the random sea state, the analysis of Miles^(2.7) provides a basis for computing the string life. Miles defines an equivalent stress of constant amplitude, σ_r which produces the same fatigue damage as the root-mean-square stress σ_{rms} induced in the string by the random sea. This equivalent stress is found by adopting the linear cumulative damage concept ("Minér's rule") and takes the form

$$\sigma_r = \sqrt{2} \left\{ \Gamma \left(1 + \frac{m}{2} \right) \right\}^{1/m} \sigma_{\rm rm\,s} \tag{2.45}$$

where Γ is the gamma function and *m* is the exponent in Eq. (2.41). In the present study, where *m* = 3.82,

$$\sigma_r = 1.662 \ \sigma_{\rm rms}$$
 (2.46)

The number of fatigue cycles to failure is, then,

$$N(\sigma) = \left[\frac{11.023 \times 10^5 \ K(1 - \sigma_s/\sigma_u)}{\sqrt{2} \ \sigma_{\rm rm\,s}}\right]^m \left\{\Gamma \left(1 + \frac{m}{2}\right)\right\}^{-1}$$
(2.47)

and the fatigue life in hours is

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$$L = \frac{2\pi N(\sigma)}{3600\omega_{\rm rm\,s}} \tag{2.48}$$

(2.49)

where $\omega_{\rm rms}$ is found via Eq. (2.31).

When encountering sea states (deterministic or random) which are not constant in time but vary, the concept of damage accumulation is necessary to determine the life of the drill string. The simplest and most useful of the several damage theories proposed is the linear cumulative damage principle associated with the names Palmgren and Miner. In this theory, a *damage ratio* is defined as the ratio of cycles experienced at a given stress level, n, to the cycles required to fail the drill pipe at that same stress level, N. As stress cycles accumulate at various stress levels, the damage ratios are added up to yield the damage fraction D:

$$D = \sum_{i} \frac{n_i}{N_i}$$

Failure is assumed to occur when the damage fraction D reaches unity, D = 1. (In practice, failures are found to occur for D as little as 0.5 and as high as 5, or more.) Thus, if the subscript *i* denotes the ith of several different stress levels and associated number of applied cycles, the cumulative damage is

$$D = \frac{1}{(11.023 \times 10^5)^{3.82}} \sum_{i} n_i \sigma_i^{3.82} = 8.47 \times 10^{-24} \sum_{i} n_i \sigma_i^{3.82}$$
(2.50)

$$D = \frac{2^{3.82/2} \Gamma[1 + (3.82/2)]}{[11.023 \times 10^5 K(1 - \sigma_s/\sigma_u)]^{3.82}} \sum_i n_i \sigma_{\rm rm\,s_i}^{3.82} = \frac{8.61 \times 10^{-2.3}}{(1 - \sigma_s/\sigma_u)^{3.82}} \sum_i n_i \sigma_{\rm rm\,s_i}^{3.82}$$
(2.51)

for the deterministic and random seas, respectively. The concept of linear cumulative damage can be very useful in keeping records of the history of drill pipe segments, thereby to estimate the remaining useful life.

SECTION 3: ROLL AND PITCH EFFECTS

As the ship rolls and pitches in response to wave action, the bending in the upper portion of the drill string must be controlled by some means to prevent string breakage due to excessive bending stresses. Bending of the drill string is also associated with the fact that the ship does not rest precisely over the drilling target. To limit the bending stresses in the string, a constraining system of some kind must be employed. This report considers the design parameters associated with two possible systems:

- (1) Guide shoe (horn)
- (2) Picalo.

Before considering each of these concepts in some detail, a few mathematical results common to both will be derived.

The drill pipe is considered as a uniform tubular Bernoulli-Euler beam, subject to tension and bending. The pipe is fitted with rubbers at a spacing s, so that only the rubbers may contact the bending constraint system. Within the usual assumptions of elementary beam theory, the deflection curve between each rubber is described by the differential equation

$$EI_p \frac{\mathrm{d}^2 y}{\mathrm{d}x^2} - Ty = M(x) \tag{3.1}$$

together with the boundary conditions

at
$$x = 0$$
: $y = 0$, $\frac{dy}{dx} = 0$ (3.2a)
at $x = s$: $y = \frac{s^2}{2R}$, $\frac{dy}{dx} \simeq \frac{dy}{ds} = \frac{s}{R}$ (3.2b)



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(3.3)

In these expressions, EI_p is the flexural rigidity of the drill pipe, R is the radius (assumed constant) to which the pipe is bent, and T is the tension and M the moment developed in the drill pipe. The other symbols are defined in Figure 3.1. The solution to this system of equation: determines the maximum bending moment as

$$M = \frac{EI_p \alpha s}{2R} \frac{\sinh \alpha s}{\cosh \alpha s - 1}$$

where $\alpha^2 = T/EI_p$.

To show this result, note that Eq. (3.1) can be expressed as

$$\frac{EI_p d^4 y}{dx^4} - \frac{T d^2 y}{dx^2} = \frac{d^2 M}{dx^2} = q = 0,$$
$$\frac{d^4 y}{dx^4} - \frac{\alpha^2 d^2 y}{dx^2} = 0$$
$$\alpha^2 = \frac{T}{EI_p}$$

or

......

The general solution is

$$y = A + Bx + C \cosh \alpha x + D \sinh \alpha x$$

Invoking the first two boundary conditions, Eqs. (3.2a), one finds

$$C = -A$$
$$D = -\frac{B}{\alpha}$$

Invoking the second two boundary conditions, Eqs. (3.2b), there results

$$A = \frac{s}{2R\alpha} \frac{\sinh \alpha s}{\cosh \alpha s - 1}$$
$$B = \frac{s}{2R}$$

From Eq. (3.1),

$$M = EI_p \frac{d^2 y}{dx^2} - Ty = -T(A + Bx)$$

and, therefore, M assumes a maximum value at x = 0, M = -TA, the magnitude of which provides the result expressed as Eq. (3.3). This bending moment acts at each rubber or tool joint.

The maximum bending stress in the drill pipe, which occurs at the rubbers or tool joints, is

$$\sigma = \frac{Mr}{I_p} = \frac{Er\alpha s}{2R} \frac{\sinh \alpha s}{\cosh \alpha s - 1}$$
(3.5)

while the peak tensile stress is

$$\sigma = \frac{T}{A} + \frac{Er\alpha s}{2R} \frac{\sinh \alpha s}{\cosh \alpha s - 1}$$
(3.6)

where A is the net area of the drill pipe section. From the form of Eq. (3.5), one sees that increasing the radius R correspondingly decreases the bending stress.

3.1 Guide Shoe (Horn) System

The Glomar Challenger is presently equipped with a guide shoe, a trumpet-shaped structure opening downward, of height h. The guide shoe has a constant radius R, to insure a uniform peak bending stress for all portions of the pipe in contact with the guide. If θ is the maximum design roll/pitch angle, then $h = R \sin \theta$, or

$$R = \frac{h}{\sin \theta}$$
(3.7)

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The maximum stress experienced by the drill string as it bends to conform to the radius of the guide shoe is given by Eq. (3.5), where R is given by Eq. (3.7). This bending stress adds to the direct tension stress T/A to become the design criterion for the string under static loading.

As the drill string rotates and decends during drilling operations, that portion in the guide shoe which conforms to the radius accumulates fatigue damage due to the bending and tension stresses. As a segment of the pipe passes through the guide shoe, it is in contact with the guide for only a portion of the time, since the roll and pitch angles oscillate about a zero mean angle. In accounting for these factors, one arrives at the following expression for the expected number of reversed bending cycles, n, an element of the pipe experiences while passing through the shoe:

$$a = \frac{0.637R\theta(60)(\text{RPM})}{\dot{x}}$$
(3.8)

where RPM is the rotational speed of the string and \dot{x} the drilling rate in feet/hour.

1

The numerical factor in this equation was derived by Vreeland^(1.2), whose derivation is reproduced below. Following this derivation, it will be shown by a different train of thought that this factor is consistent with relatively close upper and lower bound estimates. Vreeland begins by postulating harmonic roll/pitch motion of period T

$$\theta = \theta_0 \sin \frac{2\pi t}{T}$$

The drilling rate is small compared to the rate of contact of the pipe with the guide, $\dot{x} \ll R\theta_O/T$, and a segment is in contact with the guide for a time $t < R\theta_O/\dot{x}$. In following an element of the pipe as it passes downward from the top of the guide, one locates its position at any time t by $x = \dot{x}t$, and finds that it comes in contact with the guide only if $x < R\theta_O$. The element is not in contact from time t = 0 to such time when $R\theta(t) = x$, i.e., when

$$\frac{x}{R} = \theta_0 \sin \frac{2\pi t}{T}$$

Thus,

$$\frac{t_{nc}}{T} = \frac{2}{\pi} \operatorname{Sin}^{-1} \left(\frac{x}{R\theta_0} \right)$$

where $t_c + t_{nc} = T$, with the subscripts denoting "contact" and "no contact," respectively. Averaging over the length of contact, there results

$$\frac{t_c}{T} = 1 - \frac{2}{\pi R \theta_0} \int_0^{R \theta_0} \operatorname{Sin}^{-1} \left(\frac{x}{R \theta_0} \right) \mathrm{d}x = 0.637$$

which is the factor appearing in Eq. (3.8).

To justify this numerical estimate, it will now be shown that, within the assumptions presented above, the factor is bounded by 0.500 and 0.681. To do this, first follow an element of the pipe as it descends with velocity \dot{x} from the top of the guide and compare its position with the contact arc length X, Figure 3.1.1a. This comparison is illustrated in





Figure 3.1.1.b-Estimation of Time of Contact

Figure 3.1.1b. The total contact time, t_c , is found by summing up the shaded portions of the abscissa, each of which indicates the proportion of a half-cycle of roll/pitch that the element in question contacts the guide shoe. One begins by writing $t_0 = 0$

$$t_1 = \frac{R}{\dot{x}} \left| \sin \left(\theta_0 \sin \omega t_1\right) \right| = \frac{R}{\dot{x}} \sin \left(\theta_0 \sin \omega \overline{t}_1\right), \quad \overline{t}_1 = \frac{\pi}{\omega} - t_1$$

$$t_2 = \frac{R}{\dot{x}} \left| \sin \left(\theta_0 \sin \omega t_2\right) \right| = \frac{R}{\dot{x}} \sin \left(\theta_0 \sin \omega \overline{t}_2\right), \quad \overline{t}_2 = t_2 - \frac{\pi}{\omega}$$

$$t_3 = \frac{R}{\dot{x}} \left| \sin \left(\theta_0 \sin \omega t_3\right) \right| = \frac{R}{\dot{x}} \sin \left(\theta_0 \sin \omega \overline{t}_3\right), \quad \overline{t}_3 = \frac{2\pi}{\omega} - t_3$$

$$\vdots$$

$$\vdots$$

$$t_0$$

where each of the t_1 , t_2 , etc., is less than $\pi/2\omega$. Since these expressions are nonlinear in the t_1 , t_2 , etc., it is expedient to resort to bounding approximations.

For an upper-bound estimate, the bilinear approximation denoted by X' in Figure 3.1.1b is introduced: $X = R\theta_0 \omega t$, $t < \pi/2\omega$. Then, one writes

$$\begin{aligned} t_{1}^{\prime} &= 0 \\ t_{1}^{\prime} &= \frac{R\theta_{0}}{\dot{x}} \,\omega \overline{t}_{1}^{\prime} = \frac{R\theta_{0}}{\dot{x}} \,\omega \left(\frac{\pi}{\omega} - t_{1}^{\prime}\right) , \quad t_{1}^{\prime} &= \frac{\pi R\theta_{0}}{R\theta_{0}\omega + \dot{x}} \\ t_{2}^{\prime} &= \frac{R\theta_{0}}{\dot{x}} \,\omega \overline{t}_{2}^{\prime} = \frac{R\theta_{0}}{\dot{x}} \,\omega \left(t_{2}^{\prime} - \frac{\pi}{\omega}\right) , \quad t_{2}^{\prime} &= \frac{\pi R\theta_{0}}{R\theta_{0}\omega - \dot{x}} \\ t_{3}^{\prime} &= \frac{R\theta_{0}}{\dot{x}} \,\omega \overline{t}_{3}^{\prime} = \frac{R\theta_{0}}{\dot{x}} \,\omega \left(\frac{2\pi}{\omega} - t_{3}^{\prime}\right) , \quad t_{3}^{\prime} &= \frac{2\pi R\theta_{0}}{R\theta_{0}\omega + \dot{x}} \\ \vdots \end{aligned}$$

Thus,

$$t_{c}^{c} = (t_{1}^{c} - t_{0}^{c}) + (t_{3}^{c} - t_{2}^{c}) + (t_{5}^{c} - t_{4}^{c}) + \dots$$

$$= \left(\frac{\pi R \theta_{0}}{R \theta_{0} \omega + \dot{x}} - 0\right) + \left(\frac{2\pi R \theta_{0}}{R \theta_{0} \omega + \dot{x}} - \frac{\pi R \theta_{0}}{R \theta_{0} \omega - \dot{x}}\right) + \left(\frac{3\pi R \theta_{0}}{R \theta_{0} \omega + \dot{x}} - \frac{2\pi R \theta_{0}}{R \theta_{0} \omega - \dot{x}}\right) + \dots$$

$$= \frac{\pi R \theta_{0}}{R^{2} \theta_{0}^{2} \omega^{2} - \dot{x}^{2}} \left\{ R \theta_{0} \omega (1 + 1 + 1 + \dots) - \dot{x} [1 + 3 + 5 + \dots + (2n - 1)] \right\}$$

where n is here the number of half waves. Thus,

$$t'_{c} = \frac{n\pi R\theta_{0}}{R^{2}\theta_{0}^{2}\omega^{2} - \dot{x}^{2}} (R\theta_{0}\omega - n\dot{x})$$
$$\simeq \frac{n\pi}{R\theta_{0}\omega^{2}} (R\theta_{0}\omega - n\dot{x})$$

since $\dot{x} \ll R\theta_0/T$. To find *n*, note that

$$t_{\max} = n \left(\frac{\pi}{\omega}\right) = \frac{R}{\dot{x}} \sin \theta_0$$
$$\simeq \frac{R\theta_0}{\dot{x}}$$

Thus, $n = R\theta_0 \omega / \pi \dot{x}$, from which

$$t_c' = \frac{\pi - 1}{\pi} \frac{R\theta_0}{\dot{x}} = \frac{0.681R\theta_0}{\dot{x}}$$

which serves as an upper-bound estimate.

The lower-bound estimate proceeds in exactly the same manner, this time with X'', as depicted in Figure 3.1.1b. On replacing X with $X'' = (2/\pi) R \theta_0 \omega t$ and repeating the process just outlined, one concludes that

$$t_C'' = \frac{1}{2} \frac{R\theta_0}{\dot{x}} = \frac{0.5R\theta_0}{\dot{x}}$$

Inasmuch as $t_c'' < t_c < t_c'$, the estimate stated as Eq. (3.8) will be adopted in this report.

The fractional fatigue damage incurred by the element as it descends through the guide shoe is F:

$$F = n/N \tag{3.9}$$

where N is the number of cycles to failure as defined by Eq. (2.43) and n is given by Eq. (3.8). The total life of the drill string in hours is L:

$$L = N/(60)(\text{RPM}), \text{ hour's}$$
 (3.10)

3.2 Picalo System

An alternative system concept for limiting the bending radius of the drill string during roll/pitch is through the use of a picalo, or flexible guide. Conceptually, the picalo is a tapered, hollow cantilevered beam through which the drill pipe is fed during drilling. The taper is designed so that a constant radius of curvature is achieved, which depends upon the string tension.

Recall the differential Eq. (3.1) for the deflection of a beam which, in general, has the form

$$\frac{d^2}{dx_1^2} \left(EI \frac{d^2 y_1}{dx_1^2} \right) = T \frac{d^2 y_1}{dx_1^2}$$
(3.11)

The subscript 1 denotes the picalo, as described in Figure 3.2.1. For small curvatures, such as are encountered in the bending of the picalo, $d^2 y_1/dx_1^2 \simeq 1/R$, which is required to be constant. Substituting, one arrives at

$$\frac{d^2 I}{dx_1^2} = \frac{T}{E}$$
(3.12) Figure 3.2.1–5
Picalo S

whose complete solution is

$$I = \frac{Tx_1^2}{2E} + c_1 x_1 + c_2 \tag{3.13}$$

where c_1 and c_2 are constants yet to be determined.

The bending displacements of the drill pipe are governed by the differential equation

$$EI_p \frac{d^4 y_2}{dx_2^4} = T \frac{d^2 y_2}{dx_2^2}$$
(3.14)

where the subscript 2 denotes the drill pipe. The boundary conditions are

at
$$x_2 = 0$$
: $y_2 = 0$, $\frac{dy_2}{dx_2} = 0$, $\frac{d^2y_2}{dx_2^2} = \frac{M}{EI_p}$ (3.15a)

as
$$x_2 \to \infty$$
: $\frac{dy_2}{dx_2}$ is bounded (3.15b)

The solution is

$$y_2 = \frac{EI_p}{RT} (e^{-\alpha x_2} - 1) + \frac{x_2}{R\alpha}$$
(3.16)

where $\alpha = \sqrt{T/EI_p}$ as before.

To show this, one begins with the general solution of Eq. (3.14)

 $y_2 = A + Bx_2 + C \cosh \alpha x_2 + D \sinh \alpha x_2$

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 $y_{1}(x_{1})$ $y_{2}(x_{2})$ x_{2} θ



which reduces to

$$y_2 = A(1 - \cosh \alpha x_2) + B x_2 - \frac{1}{\alpha} \sinh \alpha x_2$$

under imposition of the first two boundary conditions in Eq. (3.15a). The third boundary condition provides A = -M/T. The remaining condition, Eq. (3.15b), requires bounded behavior as

$$\lim_{x_2 \to \infty} \frac{\mathrm{d}y_2}{\mathrm{d}x_2} = \lim_{x_2 \to \infty} \left[B + \frac{1}{2} \left(\frac{M\alpha}{T} - B \right) e^{\alpha x} \right]$$

From this, it follows immediately that B = MO(T), and one is led to Eq. (3.16) above.

In order to complete the stress analysis of the drill string, the constants, c_1 and c_2 in Eq. (3.13) must be determined. The bending moment M and the shear force V at the point where the drill string emerges from the lower end of the picalo are, respectively,

$$M = EI_p \frac{d^2 y_2(0)}{dx_2^2} = \frac{EI_p}{R}$$
(3.17)

$$V = -EI_p \frac{d^3 y_2(0)}{dx_2^3} = \frac{EI_p \alpha}{R}$$
(3.18)

It is necessary to maintain a smooth transition in both the bending moment and the shear in the drill string as it emerges from the end of the picalo. These two conditions suffice to determine c_1 and c_2 , and the final expression for *I*, from Eq. (3.13), is

$$I = I_p \left\{ 1 - \alpha (h - x_1) + \frac{\alpha^2 (h - x_1)^2}{2} \right\}$$
(3.19)

This result derives directly from the stated boundary conditions

$$\frac{EI}{R} \bigg|_{x_1 = h} = \frac{EI_p}{R} \qquad (\text{moment continuity})$$
$$\frac{E}{R} \frac{dI}{dx_1} \bigg|_{x_1 = h} = \frac{EI_p \alpha}{R} \qquad (\text{shear continuity})$$

where Eqs. (3.17) and (3.18) have been used. Thus, from Eq. (3.13),

$$\frac{Th^2}{2E} + c_1h + c_2 = I_p$$
$$\frac{Th}{E} + c_1 = I_p \alpha$$

from which

$$c_1 = \frac{(EI_p\alpha - Th)}{E}$$
$$c_2 = I_p(1 - \alpha h) + \frac{Th^2}{2E}$$

Using these constants, one is led to Eq. (3.19).

To compute the bending stresses at pipe elements within the picalo, the bending radius of the picalo needs to be determined in terms of the roll/pitch angle of the ship. The total roll/pitch angle, θ , is the sum of the angle φ through which the picalo bends and β , the angle through which the drill string bends: $\theta = \varphi + \beta$. The angle φ is given simply by $h \simeq R\varphi$. β is found by:

$$\beta = \lim_{\substack{x_2 \to \infty \\ x_2 \to \infty}} \frac{dy_2}{dx_2} - \frac{dy_2(\theta)}{dx_2}$$
$$= \lim_{\substack{x_2 \to \infty \\ x_2 \to \infty}} \left(-\frac{EI_p \alpha}{RT} e^{-\alpha x_2} + \frac{1}{R\alpha} \right) = \frac{1}{R\alpha}$$

Thus,

$$R = \frac{h}{\varphi} = \frac{h}{\theta - \beta} = \frac{(h + 1/\alpha)}{\theta}$$
(3.20)

The bending stress in the picalo is σ .

$$\sigma = \frac{Mc}{I} = \frac{Ec}{R} = \frac{Ec\theta}{h+1/\alpha}$$
(3.21)

where c is the outer radius of the picalo at the position in question. The bending stress in the drill string contained in the picalo is given by combining Eqs. (3.5) and $(3.20)^*$:

$$\sigma = \frac{Er\theta \, \alpha s}{2(h+1/\alpha)} \frac{\sinh \alpha s}{\cosh \alpha s - 1} \tag{3.22}$$

The maximum tensile stress is found by adding the direct tensile stress

$$\sigma = \frac{T}{A} + \frac{Er\theta\,\alpha s}{2(h+1/\alpha)} \frac{\sinh\,\alpha s}{\cosh\,\alpha s - 1}$$
(3.23)

The above equations are valid for a picalo system designed to a chosen value of tension T (i.e., a chosen α). Thus, the system is designed to some string length, and corresponding T, which insures a constant bending radius R for that string length. When drilling with a string of different length, the radius R will not be constant along the picalo, and the effective new bending radius must be estimated. The effective bending radius may be estimated by the expression

$$R = R_d + \sqrt{\frac{EI_p}{T\theta^2}} \left(1 - \sqrt{\frac{T}{T_d}} \right)$$
(3.24)

where the subscript d stands for the design condition, for which R is everywhere constant. This expression predicts $R > R_d$ whenever $T < T_d$, and vice versa. For purposes of numerical computation, Eq. (3.24) can be used in place of Eq. (3.20) for all string lengths.

Eq. (3.24) is found by requiring $h/\theta [= R - (1/0\theta)]$ to be the same for the design string length condition as for any other length. Thus,

$$R - \frac{1}{\alpha \theta} = R_d - \frac{1}{\alpha_d \theta}$$

which results in the foregoing expression.

An exact solution for the nonuniform curvature of the picalo under other than the design condition can be found, although it is tedious. One begins by writing

$$\frac{\mathrm{d}^2}{\mathrm{d}x^2} \left(I \frac{\mathrm{d}^2 y}{\mathrm{d}x^2} \right) = \frac{T}{E} \frac{\mathrm{d}^2 y}{\mathrm{d}x^2}$$

*Refer to the discussion relating to Eq. (3.24).

where $y \equiv y_1$ and $x \equiv x_1$, and I is defined by Eq. (3.19). On substituting for I, there results

$$\frac{\mathrm{d}^2 f}{\mathrm{d}x^2} + \frac{2I'}{I} \frac{\mathrm{d}f}{\mathrm{d}x} = 0$$

where $f = d^2 y/dx^2$, and the superior prime denotes d/dx. By transforming from f to g via

$$f = g \exp\left\{-\int \frac{I'}{I} \mathrm{d}x\right\}$$

the differential equation reduces to the simpler form

$$\frac{\mathrm{d}^2 g}{\mathrm{d}x^2} - \frac{I''}{I}g = 0$$

The quantity I''/I is rational, whose numerator is a constant and whose denominator is a second-order polynomial in x. By making two successive linear transformations of the independent variable,

z = ax + b $\omega = cz + d$

The differential equation can be transformed to the canonical form

$$\frac{\mathrm{d}^2 g}{\mathrm{d}\omega^2} + \left[\frac{\nu(\nu+1)}{1-\omega^2}\right]\omega = 0$$

whose solution is expressed in terms of Legendre functions

$$g = \sqrt{1 - \omega^2} \left[A P_{\nu}(\omega) + B Q_{\nu}(\omega) \right]$$

Transforming back, this result determines d^2y/dx^2 in terms of the constants A and B. Integrating twice, y is thus expressed in terms of four constants, whose values are fixed by the boundary conditions imposed on the picalo system. The final result is not given here in view of the adequacy of the approximation, Eq. (3.24).

In computing the fatigue damage inflicted in the pipe as it descends through the picalo, account must be taken of the fact that the picalo oscillates slowly in response to the roll and pitch motion of the ship. As an approximation to this effect, it is assumed that θ varies harmonically about a mean of zero: $\theta = \theta_{max} \sin(2\pi t/T)$. The effective value of θ is taken as the root mean square value

$$\theta = \frac{\sqrt{2} \theta_{\max}}{2} \tag{3.25}$$

This value of θ is to be used, for the purposes of fatigue calculations. in computing the bending stress, Eq. (3.22). This stress, in turn, is used to compute the number of cycles to failure N from Eq. (2.43). The number of cycles, n, experienced by a segment of the pipe as it descends through the picalo is

$$n = \frac{h(60)(\text{RPM})}{\dot{x}} \tag{3.26}$$

and the fraction of life expended is

$$F = \frac{n}{N}$$
(3.27)

This completes the fatigue analysis of the picalo system.

SECTION 4. COMPUTED RESULTS

This section presents data computed on the basis of the equations developed in Sections 2 and 3 for heave response and for the guide shoe and picalo concepts under conditions of roll and pitch. The data cover a wide range of parameter variations which may be encountered in the Deep Sea Drilling Project. The results are grouped into three subsections, presenting, respectively, Heave Dynamics, the guide shoe system, and the picalo system. Each subsection is prefaced by a brief commentary on the results, including major conclusions.

4.1 Heave Dynamics

Figures 4.1.1 through 4.1.22 present dynamic stress, maximum displacement, and fatigue life estimates for the drill string in response to heave motions. Most of these data correspond to sinusoidal swells, as random sea states were found to impart far less severe dynamic stresses on the string.

An important consequence of the mathematical model is that the dynamic response of the drill string is rather sensitive to the swell frequency. Thus, the ratio of wave height to wave length, which affects the swell frequency, enters as a basic program parameter in addition to wave height. The theoretical maximum sinusoidal wave height for a given wave length λ is 0.0890 λ . This value, however, is scarcely ever attained in practice and leads to dynamic stress predictions which are appreciably lower than are found with more typical wave height/wave length ratios. Ochi(2.2) shows that this ratio is commonly about 0.0286 for mild regular waves and 0.0500 for severe regular waves in Sea State 7 (very high seas). The data presented in the following curves accounted for this range of values by introducing the parameter "fraction of maximum stable sinusoidal swell" (FMSSS). The greatest permissible value of this parameter is unity, in which case the wave-height/wave-period relation is

$H_{\rm max} = 0.817T^2$

as derived in Section 2.2 of this report. The wave height/wave length which corresponds to this value is the aforementioned maximum value of 0.0890. The data which follow are plotted for three values of the fraction of maximum stable sinusoidal swell (FMSSS) of 0.1, 0.5, and 1.0. The following table shows the corresponding ratios of wave height/wave length (H/λ) and wave height/period² (H/T^2) .

H/λ	H/T^2
0.0089	0.0817
0.0445	0,4085
0.0890	0.8170
	0.0089 0.0445

TABLE 4.1. WAVE HEIGHT PARAMETERS

This range of FMSSS considered does include the estimates given by Ochi for Sea State 7. Until more definitive data are available regarding H and H/λ values actually experienced in the Glomar Challenger's drilling operations, an amount of uncertainty will exist in interpreting the data which follow. In the meantime, however, it would appear that an FMSSS of 0.5 should be reasonably representative of the heavier sea states, at least, and this fraction is therefore given somewhat more emphasis in the following curves than is either 0.1 or 1.0.



Figure 4.1.1–Heave Stress Response to Sinusoidal Swells



Figure 4.1.3–Heave Stress Response to Sinusoidal Swells



Figure 4.1.5–Heave Stress Response to Sinusoidal Swells



Figure 4.1.2–Heave Stress Response to Sinusoidal Swells



Figure 4.1.4–Heave Stress Response to Sinusoidal Swells



Figure 4.1.6–Heave Stress Response to Sinusoidal Swells



Figure 4.1.7–Heave Stress Response to Sinusoidal Swells







Figure 4.1.11-Dynamic Displacement Response to Sinusoidal Swells



Figure 4.1.8–Heave Stress Response to Sinusoidal Swells



Figure 4.1.10-Dynamic Displacement Response to Sinusoidal Swells



Figure 4.1.12–Dynamic Displacement Response to Sinusoidal Swells



Figure 4.1.13–Dynamic Displacement Response to Sinusoidal Swells



Figure 4.1.14–Dynamic Displacement Response to Sinusoidal Swells



Figure 4.1.15-Fatigue Life for Sinusoidal Swells



Figure 4.1.16-Fatigue Life for Sinusoidal Swells



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Figure 4.1.21–Effect of Percent Maximum Stable Sinusoidal Swell on Fatigue Life

Figure 4.1.22–Heave Stress Response to Random Sea States

It should be borne in mind that the fatigue life estimates are valid for a pipe section at the top of the string which supports the full string length indicated. If, in service, a pipe section is placed sometimes near the bottom of the string and sometimes nearer the top, the fatigue damage it will accumulate will be much less than if it were always near the top. The fatigue life under varying stressing conditions such as these could be evaluated using the method outlined in Section 2.5, but would require record-keeping on each strand of drill pipe. In more practical terms, if a system of rotation were developed whereby a given pipe section were placed near the bottom, centrally, and near the top of the string on a fairly regular basis, the fatigue lives of all segments would be more nearly the same and would help to forestall any premature fatigue failures.

The dynamic stress data are all given for the top of the drill string, where these stresses are maximized. The dynamic stress in the pipe is reduced uniformly all the way down the string length, and achieves its minimum at the bottom where its value is reduced from the maximum value by a factor of $\cos \alpha_1$. Using this fact, the stresses (and hence the forces) in the region of the bumper subs can easily be estimated.

All results given in this report correspond to one damping disk at the bottom of the string. In the case of a re-entry system, this disk is taken to be 16 ft in diameter to represent the landing base. Otherwise, a single damping disk 1 ft in diameter was used.

The following general observations and conclusions follow from the data which are presented in this section:

- The dynamic stresses decrease with increasing string length, for strings greater than about 14,000 ft. (The static stress, however, increases with string length, and has a counter-acting influence on the fatigue life.)
- (2) The dynamic stresses are greater for a fraction of maximum stable sinusoidal swell (FMSSS) of 0.5 than for an FMSSS of unity. Moreoever, an FMSSS of 0.5 is more nearly typical of actual values encountered.

- (3) The effect of rubber spacing on the dynamic stresses is generally very small. The 10-ft rubber spacing produces slightly lower stresses than does the 5-ft spacing. Rubber spacing is less important at low wave heights and long string lengths.
- (4) For a given string length (and FMSSS), the dynamic stress is relatively insensitive to end weight and disk diameter.
- (5) The maximum relative displacement, $v_{m ax}$ can exceed the maximum absolute displacement, $u_{m ax}$. $v_{m ax}$ ranges up to about 12 ft for long string lengths with 20-ft swells. (Note that since these are maximum values, the times at which $u_{m ax}$ and $v_{m ax}$ occur are, in general, not the same.)
- (6) The fatigue life generally remains constant or improves with increasing string length for strings longer than about 14,000 ft. [See (1) above.]
- (7) The fatigue life is decreased approximately by a factor of 10 for each 5-ft increase in wave height.
- (8) The landing base (16-ft damping disk) improves the fatigue life for string lengths greater than 14,000 ft. The improvement can be as great as 50%, depending upon the other system parameters.
- (9) The random sea state produces negligible displacements and fatigue lives that can be considered indefinite. Thus, the sinusoidal swell sea state, not the random sea state, is the governing design condition.
- (10) The response to the random sea state is insensitive to the end weight and rubber spacing, but is sensitive to the magnitude of the sea state and, to a lesser extent, to the damping disk diameter.

4.2 Guide Shoe

The following figures illustrate the type of fatigue behavior to be expected of the drill string when used in connection with the guide shoe system. Figures 4.2.1 through 4.2.7 show how the fatigue life is affected by the height of the guide shoe, and by the system parameters of angular deviation, rubber spacing, end weight, and string length. The fatigue life is expressed in terms of the ratio of the number of holes which can be drilled to the drilling rate \dot{x} . This is done for conciseness



Figure 4.2.1—Fatigue Life of String with Guide Shoe



Figure 4.1.2–Fatigue Life of String with Guide Shoe



with Guide Shoe





Figure 4.2.7–Effect of Angle of Deviation on Fatigue Life of String with Guide Shoe

inasmuch as the fatigue life is directly proportional to x. Thus, for a drilling rate of 20 ft/hr, a 50-ft guide shoe used with a 22,000-ft drill string with 5-ft rubber spacing, 40,000-lb end weight and 5° angular deviation would yield an expected life of (2.5)(20) or 50 holes.

All of the results presented for the guide shoe correspond to a design angular deviation of 8.5° . The design deviation is a critical factor in the design of the guide shoe, for it must be large enough to accommodate any conceivable roll/pitch angle (any angular deviation greater than the design value would lead to catastrophic failure due to the uncontrolled bending radius at the lower end of the shoe). On the other hand, the guide shoe radius R varies inversely with the design angular deviation θ :

$$R = \frac{h}{\sin \theta}$$

so that large values of θ imply small bending radii and, hence, large bending stresses and small fatigue lives. This tradeoff

between assured control of the bending radius and short fatigue life is the basic shortcoming of the guide shoe concept. In reviewing the accompanying figures for the guide shoe system, it should be borne in mind that the fatigue lives could be significantly improved by lessening the design angular deviation from 8.5°. *However*, clear justification for such practice would have to be provided in terms of data from severe drilling operations.

It should be recognized that the following results apply to sections of the drill string which are supporting the full indicated string length below. The lower-most sections of the string, as they descend through the guide, support very little string length, and, hence, these sections incur appreciably less fatigue damage than sections higher up on the string. This fact is made clear by studying the preceding figures. If sections of the drill string are rotated about from hole to hole by being inserted near the bottom, centrally, or upper most on the string, these sections can be expected to enjoy uniformly long fatigue lives.

The data displayed in the foregoing figures lead to the following observations concerning fatigue life of the string with the guide shoe system:

- Rubber spacing of 5 ft improves the fatigue life by a factor of about 3 over a 10-ft rubber spacing, depending upon the string length.
- (2) The fatigue life is degraded by about 30% for each additional degree of angular deviation in the range from 3 to 7°.
- (3) The fatigue life is substantially improved (nearly exponentially) with increased height of guide shoe.
- (4) The fatigue life for a 40,000-lb end weight is about 70% of the life for a zero end weight.

4.3 Picalo

Figures 4.3.1 through 4.3.8 summarize the expected fatigue life of the drill string when used in conjunction with the picalo system. The fatigue lives are presented in terms of the ratio of the number of holes which can be drilled to the drilling rate \dot{x} , just as was done for the guide shoe system. This is done for the same reason as before, viz, that the fatigue life is directly proportional to \dot{x} . Thus, these data for the picalo system can be compared directly with those for the guide shoe. All of the data presented correspond to a design length of 18,000 ft.*

Several of the conclusions drawn from the guide shoe data are applicable for the picalo system, as well. Here also, for example, it should be recognized that these fatigue lives are valid for a pipe section which is always placed so as to support a string length of the indicated amount. The fatigue lives of the string segments can be measurably increased by following a program of rotation whereby all segments are employed at the lower, central, and upper end of the string from time-to-time. If such a program is followed, the following predicted lives severely underestimate the actual lives by a rather wide margin.

The following general conclusions result from the data presented:

- (1) Rubber spacing of 5 ft improves the fatigue life of the string by a factor of from 3 to 5 over a 10-ft rubber spacing, depending on the string length.
- (2) The fatigue life of the string is degraded by a factor of about 2 to 4 for each additional degree of angular deviation in the range from 3 to 7°.
- (3) The fatigue life is improved with increased height of the picalo by a near exponential relationship.
- (4) The fatigue life for a 40,000-lb end weight is about 65% of the life for a zero end weight.

^{*}Refer to Section 3.2 of this report for a discussion of the meaning and implications of the design length.



Figure 4.3.1-Picalo Design Profiles



40,000 LB END WEIGHT

60

58

56

0 END WEIGHT

Figure 4.3.2-Bending Stresses in Picalo



Figure 4.3.3–Fatigue Life of String with Picalo System



Figure 4.3.4–Fatigue Life of String with Picalo System

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Figure 4.3.5-Fatigue Life of String with Picalo System





Figure 4.3.7–Fatigue Life of String with Picalo System



Figure 4.3.8-Fatigue Life of String with Picalo System

SECTION 5. CONCLUSIONS AND RECOMMENDATIONS

This section presents the main conclusions and recommendations which follow from the results found in the present study. These remarks, therefore, are necessarily presented in the light of the limitations of this study, viz., only two of several potentially feasible guide systems were studied in detail, and the existing structural details of the Glomar Challenger were not investigated in sufficient depth to enable any recommendations to be made regarding specific design details. The ideas of this section are presented in three separate categories: Conclusions; Recommendations for Current Operations; Recommendations Regarding Future Operations.

5.1 Conclusions

A number of detailed conclusions are presented in Section 4, Computed Results, for heave dynamics effects and for the guide shoe and picalo systems. The conclusions presented below are of a more general nature and fall within the framework of the overall objectives of the present program. Earlier sections of this report should be consulted for further information related to the topics mentioned below.

The first conclusion drawn from this study is that fatigue life, and not peak stress, is the principal design criterion for the drill string under current operating conditions. Under the most adverse set of conditions that can realistically be supposed, the peak stresses are well within the yield strength of the S-135 drill pipe.

A guide system of some kind is essential to assure the structural integrity of the string. The fatigue life of the drill string is very sensitive to the magnitude of the bending deflections which arise in string deployment and operation, and these deflections must be adequately controlled.

Since much of the work presented in this report was based upon the earlier work of Vreeland (1.1,1.2), it is appropriate to note certain comparisons between the two studies. In general, the present analysis predicts somewhat higher dynamic stresses under heave conditions than does the work of Vreeland. While there are differences in some of the mathematical relationships (in places, substantial), no unusually conservative assumptions were found in the earlier work which, if removed, would lead to increased life predictions. It was found here, as in the Vreeland study, that heave dynamics in response to random sea states is not a design condition. In fact, the present study predicts string stresses in a random sea which are somewhat lower than those predicted by Vreeland. Regarding the guide shoe and the picalo systems, the present results agree rather well with those given by Vreeland.

It may be concluded from the present study that the picalo system is much superior to the guide shoe system in terms of the fatigue life of the drill string. This conclusion follows by comparing the two under realistically severe operating conditions. Consider, for example, operations involving a 3° angular deviation in response to roll and pitch motions, and a 20,000-ft string with rubbers spaced 5 ft apart. For a drilling rate of 30 ft/hr and a 52-ft guide shoe, the string will survive about 7(30) = 210 holes, whereas with a picalo of the same length, the string will survive about 400(30) = 12,000 holes. This substantial difference is due largely to the fact that the guide shoe must be designed to accommodate angular excursions of the largest magnitude which can conceivably occur. The picalo system, on the other hand, provides bending restraint for all angular deviations, and is a much more "efficient" system.

It is useful to compare the life estimates given above with the life of the string under comparably severe heaving motions. Consider the same 20,000-ft drill string having rubbers spaced 5 ft apart. Assume a sea having 10-ft swells, and a fraction of maximum stable sinusoidal swell (FMSSS) of 0.5. The fatigue life of the string under these conditions will be about 2×10^4 hours. This will enable 1000 holes to be drilled to 600 ft each at 30 ft/hr. When compared with the fatigue life of the string with the picalo system, this calculation suggests that heave motions, not roll and pitch, may be the governing design criterion, depending, of course, on the specific picalo design. When compared with the guide shoe, roll and pitch may govern.

One popular restraining system that was not investigated in the present study is the kelly system. This system typically utilizes a specially-designed length of hex-pipe suspended from the top by a hook attached to a swivel bail. The kelly pipe passes through a gimbal-mounted bushing and attaches to the drill pipe. Hansford and Lubinski(5.1) have investigated the fatigue life of grades D and E drill pipe using a particular kelly system for restraint. They estimated that the number of holes which the pipe can survive is about 1.85 times the drilling rate, for 5° angular deviations and for a wide range of typical hook loads. This figure compares favorably with life estimates employing the guide shoe system for the present study, especially in view of the fact that S-135 drill pipe has a higher fatigue life than Grades D or E pipe. Thus, it is possible that a kelly-type restraint system would be an attractive alternative to the guide shoe or picalo. Such a system would render keelhauling unnecessary, although it makes deployment of the drill string a more time-consuming operation. Some detailed study, however, of the kelly system would have to be made in terms of the Glomar Challenger operations before meaningful comparisons could be made between these various systems.

5.2 Recommendations for Current Operations

The single major obstacle preventing more realistic estimates of the string fatigue life is the determination of the loading environments due to heave and to roll and pitch motions. Accordingly, it is recommended that procedures be effected to determine the (deterministic) sea state for typical, severe drilling environments. There are two equivalent ways of going about this:

(1) Measure the wave height H_{max} and the wave length λ directly. From these data, the period T may be calculated according to the equation

$$T^{2} = \frac{2\pi\lambda}{g} \left[1 + \left(\frac{2\pi H_{\max}}{\lambda}\right)^{2} \right]^{-1}$$
(5.1)

given in Section 2.2 of this report. Once T is known, the fraction of maximum stable sinusoidal swell (FMSSS) may be computed as

$$FMSSS = \frac{H_{max}}{0.8 \ 7T^2}$$
(5.2)

This would serve to determine the validity of the 0.5 fraction assumed for many of the results calculated in Section 4. Furthermore, the computer program could be rerun using as direct input the measured values of H_{max} and H_{max}/λ .

(2) As an alternative to the procedure outlined above, the wave height H_{max} and the period T could be measured directly. This approach would permit the FMSSS to be calculated directly from Eq. (5.2), and then compared with values suggested elsewhere in this report. In order to provide input to the computer program, the

wave length λ can be estimated from Eq. (5.1). This equation leads to a cubic expression for λ :

$$\lambda^3 - \left(\frac{gT^2}{2\pi}\right)\lambda^2 - 2\pi gH_{\max}^2 T^2 = 0$$
(5.3)

which has the solution*

$$\lambda = \frac{gT^2}{6\pi} + (A + B)$$
(5.4)

where

$$\binom{A}{B} = \sqrt{\frac{gT^2}{54}} \left(\frac{g^2 T^4}{4\pi^3} + 54 \pi H_{\max}^2\right) \pm \sqrt{\frac{g^2 T^4 H_{\max}^2}{108 \pi^2}} \left(g^2 T^4 + 108 \pi^4 H_{\max}^2\right)$$
(5.5)

Either of these two sets of measurements can be made with standard measurement tools generally available, and either procedure would lead to considerably more confidence in the fatigue life estimates of the drill string.

It is recommended that steps be taken to insure a uniform 5-ft rubber spacing on all lengths of drill pipe. While the pipe now is equipped nominally with this spacing, in fact, there is a great variability in rubber spacing due to slippage, etc. The results presented in this report show that the fatigue life of a pipe strand, with either the guide shoe or the picalo, is degraded by a factor of from three to five if there is any 10-ft spacing between adjacent rubbers. This sensitivity of fatigue life to rubber spacing is too serious to be ignored, and some program of spacing maintenance should be begun.

5.3 Recommendations Regarding Future Operations

The scope of the present study was limited specifically to evaluation of the guide shoe and picalo restraint systems. It is known that the existing guide shoe system is satisfactory in terms of assuring the integrity of the drill string under operating conditions. Desirable modifications to the present guide shoe system can only be achieved by relaxing the angular deviation design criterion (presently 8.5°), which requires a much more definitive evaluation of the roll and pitch motions of the vessel under severe drilling environments. It is possible that an entirely new guide shoe system could be designed which would have an inside diameter large enough to accommodate the landing base. Such a system would be mounted inside the moon pool. It would suffer the disadvantages in that it would protrude a significant distance below the bottom of the ship hull, would be very massive, and could not be retrieved through the moon pool.

In light of all the conclusions drawn from the present study, it is recommended that work be started on the design of a picalo restraint system. An entirely satisfactory system could be based upon a design length of less than 50 ft. Two attractive structural concepts which are compatible with ready access to the landing base and other gear at the end of the string are:

- (1) Retractable picalo
- (2) "Clamshell" picalo.

^{*}It can be shown that Eq. (5.3) has but one real root.

Of these two, the first is the more attractive if it can be designed within the constraints of the existing ship structure. This concept involves a picalo that, when in operation, is fixed at its upper end at the deck level immediately below the drilling platform and protrudes down into the moon pool. For purposes of access to the landing base and other gear, the picalo would be retracted upwards through the drilling platform to clear the area within and immediately above the moon pool. This could be done by means of an electric or hydraulic hoisting device.

The "clamshell" concept involves fixing the picalo permanently at the deck level immediately below the drilling platform. The picalo would be split vertically, so as to consist of two halves which would be bolted or otherwise fastened together during string deployment. For access purposes, the two halves would open apart at the hinge points and fold upward out of the way.

The determination of which of these two design concepts is the more attractive and technically feasible would depend upon a detailed study of the existing ship structure and its ability to accommodate a picalo structure of the size required.

If it is found that neither of the above-mentioned picalo concepts (or an equivalent) is technically feasible, it is recommended that a study of the kelly system vis-a-vis a modified guide shoe be undertaken. The present report gives all the analysis needed to evaluate a modified guide shoe system. The kelly system, on the other hand, would require some preliminary analytical study to determine whether an existing kelly would be adequate, or if a specially designed system (employing, perhaps, a Greyhound string) would be needed. Then, on the basis of a comparison between the kelly and the modified guide shoe systems, a new restraint system design would be undertaken.

SECTION 6. COMPUTER PROGRAM DOCUMENTATION

6.1 General

The program SCRIPPS calculates the stresses in the drill string and the expected string life for drilling operations under the conditions of ship heave, and roll and pitch. The program will handle a wide variety of drilling conditions including variations in the sea state, the roll or pitch angle of the ship, the drill string configuration, and the drilling rates. In addition, for roll and pitch, the calculations can be performed with either a rigid guide (guide shoe) or a flexible guide (picalo).

To calculate the stresses in the drill string for ship heave, the equations governing the longitudinal motion of the string are solved to obtain the maximum displacement and dynamic stress. A solution can be obtained for heave motions of the ship due to swells of constant amplitude and period, or to random seas. For response to swells, the sea state is defined by the wave height, length, and period. Input to the program is in terms of the double amplitude of the wave height and the wave height to wave length ratio. An expression relating the period of the wave to its height and length is included in the program. The basis for the expression is explained in Section 2.2 of this report.

Because the response of the drill string is sensitive to the swell frequency, the ratio of the wave height to wave length, which affects the swell frequency, significantly affects the string response. Hence, for any given wave height, it may be necessary to vary the wave length over some reasonable range to find the maximum string response. If the relationship of wave height to wave length is fairly constant in the open seas, then such a procedure would not be warranted. For maximum stable sinusoidal swells, the ratio of wave height to wave length is 0.0890. Ochi(2.2) gives the ratio for mild regular and severe regular waves in Sea State 7 as 0.0286 and 0.050, respectively. In the example problem, included at the end of the section, maximum stable sinusoidal swells were assumed.

For the response to random seas, the integrated spectral density of the sea (the area under the power spectral density plot) is the input parameter which defines the sea state. A trigonometric function is used by the program to approximate the shape of the power spectral density curve. Refer to Section 2.2 of the report for a description of this function.

In addition to the sea state, the longitudinal response of the string is affected by the length of the string, the concentrated mass at the end of the string, the number and size of the rubbers, and the number and size of the damping disks, all of which are included as input parameters.

For roll and pitch, the stresses in the string and the string life are determined for an assumed angular deviation of the ship in roll and/or pitch motions. These calculations can be performed for either the rigid or the flexible guide. When the flexible guide is specified, the guide stiffness parameters are calculated by the program and included in the printout. Since the flexible guide can be designed to provide constant radius for only one value of tension in the drill string, a design length must be specified which, together with the concentrated mass at the end of the string, determines the design tension load. For the rigid guide, the height of the guide radius. In addition to the angular deviation of the ship in roll or pitch and the guide design parameters, the stresses in the drill string and the string life are affected by the string length, rubber spacing, concentrated mass at the end of the string. and the drilling rate. All of these parameters are included as input data.

Standard pipe parameters are included as constants in the program, that is, parameters for a 5-in. diameter S-135 steel pipe weighing 19.5 lb/ft. In addition, an option is provided so that new pipe parameters can be inserted if desired. Other options permit the user to calculate the response of the string either to heave motion or roll and pitch of the ship, to select the type of guide to be used for roll and pitch, and the type of seas for heave response. Also, for a single run up to twenty different values of the string length, concentrated mass at the end of the string, swell height, random sea state, and angle of roll or pitch of the ship can be included. These multiple valued parameters are nested in "DO loops" in the program, as shown below:



The input variable for the outer loop is always the concentrated mass at the end of the string, and for the inner loop, the string length. For the middle loop, the input variable depends upon the type of calculation and, thus, the subroutine being used in the program. (The subroutines and their functions are explained under the program listing.)

6.2 Input Data

1.	HEADING CARD (10A8)	TT I'm a la ciata da ciata de contente
	Cols. 1-80: HED	Heading to be printed with the output.
2.	CONTROL CARD (815)	
	Cols. 1-5: IPP	Index for pipe parameter option-punch 1 if new pipe parameters are to be input; otherwise leave blank.
	6-10: IHEAVE	Index for heave response options
		blank - omit heave response
		+1 - heave response to swells
		+2 - heave response to random seas
		-1 - heave response to both swells and random seas
	11-15: IPITCH	Index for roll and pitch options
		blank - omit roll and pitch calculation
		+1 - roll and pitch with flexible guide
		+2 - roll and pitch with rigid guide
		-1 - roll and pitch with both the
		flexible and rigid guides
	16-20: MXL	Number of different string lengths to be included in the input data (20 maximum)

	21-25: MMC	Number of different values of concentrated mass at the end of the string to be included in the input data (20 maximum)
	26-30: MHMAX	Number of different wave heights to be included in the input data (for heave response to swells only) (20 maximum)
	31-35: MSTATE	Number of different random sea states to be included in the input data (for heave response to random seas only) (20 maximum)
	36-40: MTHETA	Number of different values of the angular deviation of the ship in pitch or roll to be included in the input data (for roll and pitch option only) (20 maximum)
3.	GENERAL DATA (8E10.0) Cols. 1-10: RSPACE	Spacing of the rubbers along the string (ft)
	11-20: XL(1)	Length of drill string (ft)
	21-30: XL(2)	Continue until XL(I), I = 1, MXL is satisfied
	31-40: XMC(1)	Mass at end of string (lb)
	41-50: XMC(2)	Continue until $XMC(I)$, $I = 1$, MMC is satisfied

4. ALTERNATE PIPE PARAMETERS [if IPP = 1] (6E10.0)

Cols. 1-10: DP	Outside diameter of the pipe (in.)
11-20: AP	Pipe cross-sectional area (in ²)
21-30: WTP	Weight of pipe (lb/ft)
31-40: SU	Ultimate tensile strength of the pipe (psi)
41-50: E	Elastic modulus of the pipe (psi)
51-60: DRUB	Outside diameter of the rubbers (in.)

5. DAMPING DISKS [for heave response] (3E10.0)

Cols. 1-10: XN	Number of damping disks at end of string
11-20: DD	Diameter of the damping disks (ft)
21-30: RATIO	Wave height divided by wave length (for response to swells only)

6. SWELL HEIGHTS [for heave response to swells] (8E10.0)

Cols. 1-10: HMAX(1)	Double amplitude of swell height (ft)
11-20: HMAX(2)	Continue until HMAX(1), I = 1, MHMAX is satisfied

7. RANDOM SEA STATE [for heave response to random seas] (8E10.0)

Cols. 1-10: STATE(1)	Integrated spectral energy of the random sea (ft ²)
11-20: STATE(2)	Continue until STATE(1), I = 1, MSTATE is satisfied

8. ROLL AND PITCH (6E10.0)

Co	ls. 1-10: RPM	Rotary drilling speed (rpm)
	11-20: XDOT	Linear drilling rate (ft/hr)
	21-30: H	Height of guide (ft)
	31-40: DTOP	Inside diameter at top of guide (in.)
	41-50: THETAD	Maximum angle for which the guide shoe is designed and the angle for which the stress in the flexible guide is calculated (degrees)
	51-60: XLD	Length of drill string for which the flexible guide design parameters are determined (ft)

9. ANGULAR DEVIATION IN ROLL OR PITCH (8E10.0)

Co	ols. 1-10: THETA(1)	Single amplitude angular roll or pitch (degrees)	angle of the ship	
	11-20: THETA(2)	Continue until THETA(1), I = 1, MTHE	TA is satisfied	

6.3 Output Information

- 1. HEADING-exactly as input on Card 1
- 2. HEAVE RESPONSE
 - a. Heading to identify the type of response, i.e., heave response to swells or random seas
 - b. Input data which remain invariant for the run, i.e., the rubber spacing, the number and diameter of the damping disks, the wave height to wave length ratio and the fraction of max, stable sinusoidal swells
 - c. Subheading to identify concentrated mass at end of string and the swell height or random sea state

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- d. For each end mass and sea state combination, the following quantities are printed for each string length [for random seas, quantities (1) and (6) are rms value]:
 - (1) Maximum dynamic stress [at top of string] (psi)
 - (2) Equivalent stress which provides for the effects on fatigue life of surface corrosion by the sea water and the mean stress level in the pipe (psi)
 - (3) Maximum static stress in the string which is equal to the string weight less buoyancy effects (psi)
 - (4) Cycles to failure of the string for continuous cycling at the maximum stress levels
 - (5) Life of the string for continuous operation at the maximum stress levels (hours)
 - (6) Maximum displacement of the end of the string relative to the ship (ft)
 - (7) Maximum absolute displacement of the end of the string [response to swells only] (ft)

3. ROLL AND PITCH

- a. Heading to identify type of guide, i.e., roll and pitch-rigid guide or flexible guide
- b. Input data which remain invariant for the run, i.e., the rubber spacing, height of guide, maximum design angle (for rigid guide), design length (for flexible guide), the rotary drilling rate and the linear drilling rate
- c. If the flexible guide is used, the outside diameter and moment of inertia of the guide are printed for ten stations along its length. In addition, the maximum stress in the guide (at the top) for an angular deviation of THETAD and with tension in the string corresponding to the design length, XLD, is printed.
- d. Subheading identifying the end mass on the string and the roll or pitch angle of the ship
- e. For each combination of end mass and roll or pitch angle, the following quantities are printed for each string length:
 - Maximum dynamic stress in string [bending stress in the pipe while inside the guide] (psi)
 - (2) Equivalent dynamic stress which provides for the effect upon the fatigue life of surface corrosion by the sea water and the mean stress level (psi)
 - (3) Maximum static stress in the string which is equal to the string weight less buoyancy effects (psi)
 - (4) Cycles to failure of the pipe for continuous cycling at the maximum stress levels

- (5) Life in hours-(4) converted to hours based on the rotary drilling rate
- (6) Number of holes which can be drilled under the stated conditions-accounts for variations in the stress as ship rolls or pitches and the drilling rates

6.4 List of Program Parameters

Fortran Symbol	Definition	Variable Name
1. INPUT PARAMETERS		
a. General		
RSPACE	Rubber spacing along the string (ft)	s
XL	String length (ft)	R
ХМС	Mass at end of string (lb)	М
b. Optional Pipe Param	eters	
AP	Pipe cross-sectional area (in ²)	Ap
DP	Pipe outside diameter (in.)	dp
DRUB	Outside diameter of rubbers (in.)	dr
Е	Elastic modulus of pipe (psi)	Е
SU	Ultimate tensile strength of pipe (psi)	Su
WTP	Weight of the pipe (lb/ft)	(µ)(g)
c. Heave Response		- 4 E
DD	Diameter of the damping disks (in.)	d _d
HMAX	Double amplitude of wave height (ft)	H _{max}
RATIO	Ratio of wave height to wave length (for swells only)	Η/λ
STATE	Integrated spectral energy of the random sea (ft ²)	$\Phi_{ss}(\omega)$
XN	Number of damping disks at end of string	n

Fortran Symbol	Definition	Variable Name
d. Roll and Pitch		
DTOP	Inside diameter at top of guide (in.)	d
Н	Height of guide (ft)	h
RPM	Rotary drilling rate (rpm)	RPM
THETA	Roll or pitch angle of the ship (degrees)	θ
THETAD	Maximum angle for which the rigid guide is designed (degrees)	θ_{d}
XDOT	Linear drilling rate (ft/hr)	ż
XLD	Length for which the flexible guide is designed (ft)	ℓ _d
. PROGRAM CONSTANTS	5	
AP	Pipe cross-sectional area = 5.66 in ²	A _p
DP	Outside diameter of pipe = 5.00 in^2	dp
DRUB	Outside diameter of rubbers = 7.00 in.	d _r
CD	Coefficient of fluid drag = 1.12	CD
E	Elastic modulus of pipe = 30×10^6 psi	E
EXPM	Constant in fatigue life relationship = 3.82	m
G	Acceleration due to gravity = 386 in./sec ²	g
PI	Universal constant = 3.1416	π
RHØ	Density of sea water = 63.99 lb/ft^3	ρ _f
SU	Ultimate tensile strength of pipe = 160,000 psi	Su
WTP	Weight of pipe = 19.5 lb/ft	(µ)(g)
ХК	Constant in added mass term = $2/\pi$	$2/\pi$

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	Fortran Symbol	Definition	Variable Name
	XKS	Factor to account for effects of surface corrosion on fatigue life of pipe = 0.9	К
	ХО	Assumed displacement at end of string for initial damping calcu- lation = 50 in.	Xo
	XNU	Viscosity of sea water = 1.129 lb-sec/ft ²	v .
3. CAL	CULATED QUANTIT	TIES	
	ALFA	Equation (2.17)	α
	ALFA1	Equation (2.10)	α1
	BETA	Equation (2.11)	β
	С	Concentrated damping at end of string due to damping disks-Table 2.4.1	C
	СО	Equation (2.14)	Co
	CYC	Fatigue life of string in cycles	N
	C1	Equation (2.9)	c ₁
	D	Equivalent diameter of the rubbers	d
	DAMP	Distributed damping due to fluid drag on pipe and rubbers-Table 2.4.1	D
	DBAR	Equivalent diameter of the damping disks	d
	DIP	Inside diameter of the pipe	d
	ETA	Equation (2.32)	η
	F	Equation (2.8)	F
	GAMMAF	Gamma function	$\Gamma(1 + m/2)$
	GAMA	Equation (2.12)	γ
	HOLES	Number of holes which can be drilled under specified drilling conditions	1/F

	Fortran Symbol	Definition	Variable Name
	HSSQ	Equation (2.28)	H _S ²
	HV	Transfer function of the ship	$H_{\mathbf{V}}$
	LIFE	Life of the string in hours for con- tinuous cycling at maximum stress	L
	NN	Equation (2.7)	N
	ØMEGA	Frequency of the forcing function (sea)	Ω
	ØMEGA1	Fundamental frequency of the string longitudinal response	ω_1
	PHEWD	Equation (2.26)	$\Phi_{\eta\eta\dot{w}}$
	PHEWDD	Equation (2.27)	$\Phi_{\eta\eta\ddot{w}}$
	PHISS	Power spectral density of the sea- Equation (2.33)	Φ_{ss}
F	RDES	Radius of curvature of the flexible guide for $\theta = 8.5^{\circ}$ and design tension in string	R
	RHØRN	Radius of the guide shoe	R
	SEQ	Equivalent stress in pipe	σ _{eq}
	SGMAX	Maximum stress in flexible produced by RDES	σ
	SRMS	Dynamic stress based on rms value of ship roll or pitch angle	σ _{rm s}
	SS	Static stress in string	σ
	STRES	Maximum dynamic stress in string	σ
	STRESR	Equivalent stress for calculating fatigue life due to random seas	σ _{eq}
	Т	Period of ocean waves	Т
	TDES	Design tension in string for flexible guide	T _d
	TEN	Tension in string	Т
Fortran Symbol	Definition	Variable Name	
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TRMS	Rms value of theta	$\theta_{\rm rms}$	
T3	Equation (2.15)	Τl	
T4	Equation (2.16)	T2	
UL	Absolute displacement at end of string	u(ℓ)	
VL	Displacement of string relative to the ship	v(୧)	
WMAX	Heave motion of the ship	W	
XDG	Distance along guide measured from bottom	x, x ₁	
XIG	Moment of inertia of the guide	1	
ХМ	Total mass at end of string including added mass due to the damping disks	М	
XMEAN	Mean displacement of string for calculating distributed damping		
XMU	Density of the pipe including apparent mass effects due to rubbers		
XMUP	Density of the pipe	μ	
XN2	Number of rubbers per unit length of string	n ₂	

6.5 Program Listing

The program is composed of a main calling program and six subroutines. A brief explanation of the main program and subroutines is given below, followed by the program listing:

- SCRIPPS Main or calling program. Reads heading card. control card and the common data cards. Prints heading and calls subroutines as directed by the control card data.
- PAR Subroutine to calculate the string behavior for roll and pitch of the ship. Includes provisions for a rigid or flexible guide.
- SUB Subroutine to calculate fatigue life of the string for ship roll or pitch and print output for each string length.
- HEAVE Subroutine to calculate the response of the string to heave motions of the ship.

- COMUN Subroutine which contains statements common to the response calculations for both swells and random seas in the HEAVE subroutine.
- FATIG Subroutine to calculate the fatigue life of the string for dynamic stresses due to heave motion of the ship.
- GMMMA Subroutine to calculate the gamma function for different values of "m" other than that included as a constant in the program.

6.6 Example Problem

The example problem uses every option and, therefore, contains, every input parameter in the program. For the multivalued parameters, two each have been included, although any number from 1 to 20 could have been used. Note, however, that if the maximum number of each multivalued parameter is used, particularly in the outer loops, the output becomes exceedingly large.

The input parameters used in the example problem were selected to represent a possible drilling condition with the Glomar Challenger. The ratio of the wave height to wave length is somewhat arbitrary and represents maximum stable sinusoidal swells. This condition may not produce maximum response in drill string. (See Section 4.1 for a more detailed discussion of this point.)

The input data are shown on a data sheet exactly as they would be punched in the data cards. The computer printout of the results follows the input data.

2	4 5	6 7 8	9 10	11 12	13	14 15	16	17 1	8 1	20	21	22	13 2	4 25	26	27	28	29	30	31	32	33	34	35	36	37	38	39	40	41	42	43	44	45	46	17	48	49 5	0	51 5	52 5	53 5	4 5	5 5	56 5	17 3	8 5	19 6	00	16	2 6	3 6	4	5 6	6 4	67	68	69	70	71	72	73	74	75	76	17	78
								1	T	t			I	T									t	t	t					T	t	t	t			1	1	1	1		1	1	t	t	t	1	T	T	t	T	T	T	T	T	T	T									1	1	
																								Γ															Ι					Ι	Ι	Ι		I	Ι	I	I	I	I														
X	MP	LE	PR	01	34	EN	1-	-	es	P	4	C	E			5	_	F	Z	,		D	D	1	a		1		F	1	1,		X	D	0	T		=	1	3	0	-	F		4	#	R	,	1	H		-		5	1		7	3		F	T				4	1	1
H	++-		++	++	\square	-	H	+	+	+	-		+	+			_	-	4	_		L		+	L	_	_	_		L		+	╞	-	Ц	+	4	+	4	+	+	+	+	+	+	+	+	-	4	+	+	+	4	+	+	4	Н	H	Ц	μ	Н	\square	μ	-	+	+	-
H	PP	4	IEA 4	1	EPI	TCH	+	+	4	-	-	H	1	ma	-	~	IH	_	×	_	M	57	ŕ	te		M	м			┝	+	+	╞	-	Н	4	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	+	Н	H	Н	H	μ	μ	Η	-	+	+	-
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H	+1-	AREC		++	H	51		-	-	+	┝	H	+	+	┝	-	-		Н	-	-	┝	╀	ł	╀	┝	+	-	-	┝	+	+	╀	┝	+	-	-	-	+	-	H	+	-	+	+	-	4	+	+	+	+	+	+	+	-	Н	+	H	μ	⊢	۲	+	۲	-	Н	-	┝
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PRCG:A* SCHIPPS (INPUT, OUTPUT, TAPE1=INPUT) COM***:/DATA/CP,AP,X*CP,DHUB,E.SU;XKS,EXPM,RHO,XNU,G,PI;XK*CC,IMEAV 1E,IPITC+:XL(20),XMC(20).MXL,MMC,MM*AX,MSTATE.MTHETA,RSPACE,XO COMMON ROPM(210) 000003 000003 DIME STO HEAD(4) 000003 C DATA DP, #P, #TP/05 UB/5:00.5.66/19.50.7.00/ DATA E.S. #X551 x57/30.66/100000.75.9.3.62/ DATA R-0.X U/63.9911.129/ DATA S.P1/364.75.1416/ DATA X4.00.X0 /0.63711.12.50.0/ 000003 000003 000003 000003 000 PARA ETER CONVERSIONS FOR PROGRAM CONSTANTS XNU = X'U = 32.17 = 144./RHO RHC = 4~0/(12, ==3.) 1 FORMAT(5=17) 000003 000005 000012 1 FORALT(512) 2 FORALT(513) 4 FORALT(513) 4 FORALT(513) 5 FORALT(513) 940 FORALT(513) 1 33H PIPE DIAMETER, IN, F12,2/ 1 33H RENSESSECTIONAL AREA, SQ. IN. = F12,2/ 3 33H WEIGHT PER-FOOT, LB, F12,2/ 4 33H TENSILE STRENGTH, PSI = F12,2/ 5 33H ELASTIC MODULE, PSI = F12,2/ 6 33H DIAMETER OF RUBBER, IN = F12,2/ 950 FORMAT(//44H STANDARD DRILL PIPE, 5 IN, DIA - 19,5 LB/PT) 000012 000012 000012 000012 000012 000012 ¢ 000012 6 CONTINUE READ 1, HEAD IF(EDF:1)55,7 000012 000020 TELEVITION CONTINUE PRINT 2.4EAD READ 3. IPP, IHEAVE, IPITCH, MXL, NMC, MHMAX, MSTATE, MTHETA READ 4.RSPACE, (XL(1), I=1, MXL), (XMC(1), I=1, MMC) 000023 000031 000 ARE NE- PIPE PARAMETERS WANTED IF(1PP)10,10,9 000100 IF(1PP)10109 9 CONTINUE READ 5. DP.AP.WTP.SU.E.DRUB PRINT 947, DP.AP.WTP.SU.E.DRUB GO TO 11 10 PRINT 950 11 CONTINUE 000102 000102 000122 000142 0001-3 С 000147 XMUP = .TP/(12,=4P) C WHIC- SUAROUTINES ARE WANTED CC IF(1HEAVE)20,30,15 000151 000153 15 CONTINUE IF(IHEAVE.GT.1)IHEAVE=0 20 CONTINUE CALL HEAVE 000156 000156

000157	30	CONTINUE
000157		IF(1P1TC-)40,50,35
000161	35	CONTINUE
000161		IF(IPITCH. ST. 1) IPITCH=0
000164	40	CONTINUE
000164		CALL PAR
000165	50	CONTINUE
000165		GO TO A
000166	55	CONTINUE
000166		STOP
000170	1.4	END
	C	

SUBROUTI'E PAR

	SUBROUTI E PAR
	C
000002	DIMENSION THETA(20)
000002	COMMCN/DATS/DP.AP, XMUP, DRUB, E, SU, XKS, EXPM, RHO, XNU, G, PI, XK, CD, IHEAV
000000	1E, IPITCH, XL (20), XMC (20), MXL, MMC, MHMAX, MSTATE, MTHETA, RSPACE, XC
000002	
000002	COMMON SPAC(180),XG(10),XG(10),5G(10)
	c
	C PARAMETER CONVERSIONS FOR PROGRAM CONSTANTS
000002	REAL LIFF
200000	OPI = SORT(OP=02, + 4, -AP/PI)
000013	XIP = (P1/64,)*(DP**4 = DP1**4)
	c
000020	1111 FORMAT(1-1)
000020	PRINT 1111
000024	1010 FORMAT(6E12.0)
000024	1012 FORMAT(8E10.0)
000024	1020 FORMAT(//JOH ROLL AND PITCH - RIGID GUIDE /
	1 22H HEIGTH OF GUIDE, FT =F6.2/
	2 19H GUIDE DESIGNED FOR, F6, 2, 9H DEGREES)
000024	1051 FORMAT(///36H TOTAL WEIGHT AT END OF STRING, LB =,F10,2)
000024	1052 FORMAT(//36H ANGULAR DEVIATION, DEG = F10.2)
000024	1030 FORMAT(//4x,16H STRING LENGTH ,16H BENDING STRESS ,
	1 16H EQULY, STRESS , 16H STATIC STRESS ,
	2 16H LIFE IN CYCLES , 16H LIFE IN HOURS ,
	3 16H LUMBER OF HOLES/)
000024	1060 FORMAT(//32H ROLL AND PITCH - FLEXIBLE GUIDE/
000024	
	1 22H HEIGTH OF GUIDE, FT #F6.27 2 19H GUIDE DESIGNED FOR,F10,2,11H FT STRING)
000024	1070 FORMAT(/33+ FLEXIBLE GUIDE DESIGN PARAMETERS/6X,
000024	
	1 18H DIST, FROM TIP ,18H OUTSIDE DIAMETER ,
	2 18- YOMENT OF INERTIA /10(3F18,2/)
	3 45H "AX STRESS IN GUIDE FOR ANGULAR DEVIATION OF, F6.2.
112121212121212	4 10H DEGREES =, F10, 2, 4H P51//)
000024	2000 FORMAT(/JOH ROTARY DRILLING RATE, RPM =, F10.2/
	1 30H LINEAR DRILLING RATE, FT/HR =, F10.2/
	2 JOH RUBBER SPACING. FT =, P10,2)
	C
000024	READ 1010, PP", XDOT, H, DTOP, THETAD, XLD
000044	PEAD 1012; (T-E7A(1), I=1, MTHETA)
000057	1F(1P1TCH)97,98,99
000061	97 CONTINUE
000061	PRINT 1020, H, THETAD
000071	PRINT 1060, H, XLD
000101	
	PRINT 2000, RPM, XDOT, RSPACE
000113	GO TO 100
000114	98 PRINT 1020.H. THETAD
000124	PRINT 2000, RPM, XDOT, RSPACE
000136	GO TO 100
000137	99 PRINT 1360.H. XL
000147	PRINT 2000, RPM, XOOT, RSPACE
000161	100 CONTINUE
	C
	C PARAMETER CONVERSIONS FOR INPUT DATA
	c c c c c c c c c c c c c c c c c c c
000161	наче 12.
000163	
000103	xLD = xL3+12.

000164 000166 000170	THET4D = THETAD = P1/180. Do 1900 I = 1,MMC PRINT 1031, XMC(I)
0 00	DESIGN PARAMETERS FOR FLEXIBLE GUIDE
C	
000176	TDES=X"C(1)+(X"UP=RHO)/XMUP+(XMUP=RHO)+AP+ XLD
000204	ALFAD=SORT(TDES/(E=X1P))
000210	RUES=(++1./ALFAC)/THETAD
000214	ELEM = 0.0
000215	DX=H/12.
000217	70 35 <=1,10
000220	ELEM . ELEM + DX
000222	XG(K) · ELEM
000224	XIG(K)=TCES + XG(K)+2 /(2.+E) + X1P+ALFAD+ XG(K)
000234	$\Im G(K) = (64, * X1G(K)/P1 + (DTOP)**4)**,25$
000245	35 CONTINUE
000247	SGMAX=E+DG(10)/(2.+RDES)
000252	THETAP = THETAD+183,/PI
000255	PRINT 1070, (XG(K),DG(K),XIG(K),K=1,10),THETAP,SGMAX
000301	00 1000 J = 1, "THETA
000303	PRINT 1352, THETA(J)
000311	THETA1= THETA(J) + PI/180.
000315	PRINT 1030
000320	D0 1707 H # 1, MXL

	ĉ	PRESAN CALCULATIONS
	C	
000322		TEN = x (1) (X UP - RHC)/X UP + (X UP - RHC) AP XL(K) 12.
000332		ALFAT SURT(IEN/(F + XIP))
000337		X = ALFA + RSPACE #12.
000342		SINHX = (EYP(X) - EXP(-X))/2
000351		COSHY = (EXP(X) + EXP(-X))/2
000361		IF(IPITC-)20,20,30
	C	
	000	AIGIE SUIDE
	C	
000363	20	RHOR' = -/SI'(THETAD)
000366		IF(IPITCH.LT.C) PRINT 1054
000374	1054	FORMAT(/18H RIGID GUIDE)
000374		XMOM = E + XIP + X + SINHX/(2.+RHORN+(COSHX-1.))
000403		STRES = XMOMODP/(XIPO2.)
000405		XNCYC = .637eRHORN#THET41#(RPM#60,)/(XDOT#12.)
000414		DBOT = 2,=RHORN=(1,=COS(THETAD))+DTOP
000423		TEST = J
	C	the second se
000424		CALL SUB(TEN, AP, STRES, XKS, SU, EXPM, RPM, XNCYC, XL(K), SRMS, NTEST)
	C	
000441		IF(IPITC+)30,1000,30
000442	30	CONTINUE
000442		IF(IPITCH.LT.O) PRINT 1053
000447		FORMAT(/16- FLEXIBLE GUIDE)
	ĉ	
	c	FLEXIBLE GUIDE
10120000000000	C	
000447		RERDESOTHETAD/THETA1 + SGRT(EOXIP/(TENOTHETA1002))+(1 SGRT(TEN/
		1 TDES))
000467		XNCYC = H=RPM+60./(XDOT=12.)

000474		XMOM=E=X!P= x =SI\HX/(2.+R=(COSHX=1.))
000503	520	STRES=XMCMeDP/(XIP#2.)
	C	
	С	BASE FATIGIE LIFE ON RMS VALUE OF THETA
000505		TRMS = THETA10,707
000507		RRMS#RCES#THETAD/TRMS+ SGRT(E+XIP/(TEN#THETA1++2))+(1, - SGRT(TEN/ 1 TDES))
000527		RMSMCM = EexIPexeSINHX/(2, BRRMS+(COSHX = 1,))
000536		SRMS = R*S*0**0P/(XIP+2.)
	c	
000541		NTEST = 1
	C	
000542		CALL SUB(TEN, AP, STRES, XK8, SU, EXPM, RPM, XNCYC, XL(K), SRM6, NTEST)
	C	
000560	1000	CONTINUE
000570		RETURN
000570		END

	SUBRE UT: E S.B(TE., AP, STRES, XKS, SU, EXPM, RPM, X4CYC, XL, SRMS, NTEST)
	c
	C SURRIUTI E TO CALCULATE STRESS AND FATIGUE
000016	REAL LIFE
000016	35 = TEN/4F
000017	IF(55, GT, Su)50 TO 7000
000023	SE3 = STALS / (X < S = (1, -SS/SU))
000025	IF(5.G.LT.)SEG=SEQ
000030	CYC = (11"2.3E3/SE3)**EXPM
000034	LIFE = $(\nabla C/(\partial P^{N+6}G_{*}))$
000037	IF(\TEST) 50, 107, 50
000041	50 SEGR 5 = SAMS/(XXS*(1 55/SU))
000045	IF (SF 34'S .LT. 0) SENA'S = -SE 38'S
000050	CYCR * 5 = (1102, 3E3/SE3RM\$) EXPM
000054	HOLES = CYCR S/X CYC
000057	SC TC 7032
000057	100 HOLES = CYC/X CYC
000061	50 TT 7972
2000062	7300 CONTINUE
000062	SEG =J
000063	CYC = C
000063	LIFE = 0
000064	HOLES = C
000066	PRINT 7011
000071	7001 FORMAT(64H IF SE3, CYC AND LIFE = 0 , STATIC STRESS IN STRING EXCEE
	1DS YIELD)
000371	7022 CONTINUE
	S
000071	PRINT 1550, XL, STRES, SE3, SS, CYC, LIFE, HOLES
000116	1353 FORMAT(4F14, 2, 2216, 4, F16, 1)
000116	PETUIN
000117	END
	c

		SUBROUTI & HEAVE
000002	С	DE 41 - VA
000002		REAL NY COMMENTERTINEP, AP, XMUP, DRUB, E, SU, XKS, EXPM, RHO, XNU, G, PI, XK, CD, IHEAV 15, IPITCH, XL (20), XMC(20), MXL, MMC, MMAX, MSTATE, MTHETA, RSPACE, XO
000002		COMMC, JMAX(20), STATE(20), STRES(20), SEG(20), SS(20), CYC(20), LIF 1E(20), VL(20), UL(20), SPAC(30)
000002		400 FORMAT(3F11, 2)
000002		405 FORMAT(//254 HEAVE RESPONSE TO SWELLS//) 406 FORM/T(424 RUBBER SPACING, FT =,F10,2/
		1 42H NO, OF DAMPING DISKS =,F10.2/
000002		2 424 DIAMETER OF DAMPING DISKS, FT =,F10,2) 403 FORMAT(42H RATIO OF WAVE MEIGHT TO WAVE LENGTH =,F10,2/
000002		1 42H FRACTION OF MAX STABLE SINUSOIDAL SWELL =, F10, 2) 410 FORMAT(8E10, 0)
000002		415 FORMAT(/30- AVE HEIGHT, FT =, F10.2/
000002		1 304 HEIGHT AT END OF STRING, LB #,F1D.2) 420 Format(//304 HEAVE RESPONSE TO RANDOM SEAS//)
000002		425 FORMIT(//4x,16H STRING LENGTH ,16H RMS STRESS ,
		1 14H EQUIV STRESS ,16H STATIC STRESS , 2 14H LIFE IN CYC ,16H LIFE IN HOURS ,
		3 194 RMS REL DISP, FT/(4F16,2,2E17,4,F16,2))
000002		435 FORMAT(/30- SEA STATE; FY SOD =;F10.2/ 1 3)H REIGHT AT END OF STRING, LB =;F10.2)
000002		440 FORMAT(//,4X,15H STRING LENGTH ,15H DYN STRESS ,
		1 154 FOULV STRESS , 15H STATIC STRESS , 2 15H LIFE IN CYC , 19H LIFE IN HOURS , 15H REL DIBP, PT ,
		3 15- A95 UISP, FT /(4F15,2,2E15.4,2F15.2))
000002	С	READ 400, XN, DD, RATIO
	C	PAPATETE CONVERSIONS FOR PROGRAM CONSTANTS
000014		PERCY = RATIO/0,0890 XN2 = 1,/ (RSPAGE = 12,)
000023		5= SGRT(CR.B**2-DP**2)
000025		DBAR=S34T((DD+12,)++2 =DP++2) XMU = X/(P + X ¹ /2 + XK + RHO + P1 + D++3/(6,+AP)
000045	с	CO = Sunt(E + G/XMU)
000052		IF(I=EAv-)10, 10C, 10
	000	RESFONSE TO SHELLS
000054	C	10 PRINT 405
000060		PRINT 4C4, RSPACE, XN, DD
000072		PRINT 404, RATIO, PERCNT READ 41°, (HMAX(I),1=1,MMMAX)
000115		DO SC I=1, MC
000117		XM = XMC(I) + XN + XK + RHO + PI + DB4R++3/6 D0 80 J = 1, MHMAX
000132		T= SGRT(HMAX(J)/(PERCNT+,817)) OMEGA = 2, • PI/T
000142		HV = .03 + (7 - 2.)
000144		WMAX = HVGHMAX(J)=12. Do 67 K = 1,7XL
	С	
000152	c	CALL COVEN(BETA, ALFA1, OMEGA1, C1, XM, XMU, AP, XL(K), CO)
	1.1	
000165		T3 = XM=+MAX = (C1 = OMEGA)==2 = ((1,=COS(ALFA1))/ ALFA1 = BETA 1 = SIN(ALFA1))= SIN(ALFA1)/ BETA
000206		26 XMEA* = Y0/1.414 DAMP = 1.414 = PI = (DP/2.) = RHO = SORT(XNU = OMEGA1) +,5 = XN2
		1 • CD• RHC • (PI • D•+2./4.) • X"EAN • OMEGA1
000235		C =.5 * XN * CD * RHO * (PI * DBAR**2,/4,) * XO * OMEGA1 CAMA = DAMP * XL(K) * 12./C
000254		F = WMAX = C1 = SGRT(XM ==2.= OMEGA==4. = ((1.=COS(ALFA1))/ALFA1=
		18ETA = SIV(ALFA1))=+2./BETA++2. + C ++2. + OMEGA++2.+((1CO5(ALFA 21)) = GA*A/ALFA1 + SIN(ALFA1))=+2.)
000335		"N = F/SORT((XM + OMEGA1++2 + C1++2 + (,5 + SIN(2,+ALFA1)/(4,+ALFA1)) 1)/BETA = OMEGA++2 /BETA)++2 +OMEGA++2+C1++4 + C++2 +((2,+SIN(2,+ALFA1)))
		241)/ ALFA1)+GAMA/4 + (SIN(ALFA1))++2)++2)
000401		VL(K) = 'N = C1 = SIN(ALFA1)
000407		T4 = C+_MAX + C1++2 + CMEGA + (GAMA + (1.+COS(ALFA1))/ALFA1 + SIN 1(ALFA1)) + SIN(ALFA1)
000430		ALFA = ASI'('N + C + C1++2 + OMECA + (GAMA +(2, -SIN(2, + ALFA1)) 1/ ALFA1)/4, + (SIN(ALFA1))++2)/F)
000454		(_(K) = S2PT((+MAX + N) + (T3 + COS(ALFA) = T4 + SIN(ALFA))/F) 1++2 + NN++2+(T3+SIN(ALFA) + T4+COS(ALFA))++2 /F++2)
000510		DIFF = ABS(UL(*)=XO)
000513		1F(D1FF.LE.UL(K)/200.) GO TO 50 X0 = UL(K)
000521		GO TO 27
000522		50 VL(K) = VL(K)/12. UL(K) = UL(K)/12.
000527	4	STRES(K) = (F+N,+C1+4LFA1/(XL(K)+12.))
	C	

/140

54

000537	CALL FATIG(STRES(K),SS(K),SEQ(K),CYC(K),LIFE(K),XMC(1),XMUP,RH0,XL 1(K),AP,XXS,SU,EXPM,P1,OMEGA1)
	c
000563	63 CONTINUE
000566	PRINT 414, 444X(J),X*C(1)
000577	PRINT 44", (XL(K), STRES(K), SEG(K), SS(K), CYC(K), LIFE(K), VL(K), JL(K
000377	
	1) .<=1, ×L1
000637	BO CONTINUE
	C RESFOLSE TO RANDOM SEAS
	C RESECTE TO RANDOR SEAS
	c
000644	IF(I-EAVE) 100,100,200
000646	100 PRINT 420
000652	PRINT 404, RSPACE, XN, DD
000664	PEAD \$1",(STATE(1),I=1, STATE)
000677	00 190 1 = 1. MC
000701	XM = X C(1) + X + XK + PHO + P1 + D94R++3 /6.
000712	DO 190 J=1, STATE
000713	L = STATF(J) / 1.20
000715	N = 4]
000716	NM1 = N+ 1
000720	DXOMEG = . #/N
000723	DO 142 4 = 1, MXL
	c
000725	CALL CONVIBETA, ALFA1, OMEGA1, C1, XM, XMU, AP, XL(K), CO)
	C
000740	140 XMEAN = X0/1.414
000742	DAMP = 1.414 + PI + (DP/2.)+RH0+SGRT(XNU+OMEGA1)+.5+ XN2+CD+RH0+
	1(PI = 0==2/4,)= XMEAN = OMEGA1
000765	C= .5 * XN * CD * RHO * (P1 * DBAR **2 /4,)* XO * OMEGA1
	그렇게

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000776	GAMA . DAMP . XL(K) . 12./C
001002	OMEGA = ,2
001004	SUM = C.2
001005	SUM2 . 1,1
001005	XING = 2.0
001007	COF = 2.0
001010	
	00 160 LE1, NM1
001011	COF = COF + XINC
001013	OMEGA = CHEGA + DXOMEG
001015	T = 2. • PI/OMEGA
001017	HV = .04 + (T=2)
001022	PHISS = A*(1 COS(5.*PI*(OMEGA2)/2.))*=2
001034	PHIN" = CMEGA ++2 +PHISS + HV++2
001037	PHINDD = OMEGANNO PHIND
001041	HSSG = 1/((X4=0MEGA1==2=C1==2=(.5 +SIN(2=ALFA1)/(4=ALFA1))/BETA =
	10MEGA++2/BETA)++2+C++2+0MEGA++2+C1++4+(GAMA/4+(2-SIN(2+ALFA1)/ALFA
	21)+S14(4LF41)==2)==2)
001107	PHEWD = (C1=C)=+2 =((1,-CO8(ALFA1))=GAMA/ALFA1 + SIN(ALFA1))==2+H5
	15geP+1*)
001123	PHE-UD = (C1+ XM)+=2 +((1,+COS(ALFA1))/ALFA1+BETA=SIV(ALFA1))+=2 +
001222	1HSSG + PHI-DD/BETA++2
001141	SUM = SUM + COF + (PHEWD + PHEWDD)
001144	SUM2 =5: 2 + OMEGA**2 =SUM
001147	XINC =-XINC
001150	160 CONTINUE
001153	ETA = SORT(DXOMEG = SUM/3,)
001157	VL(K) = FTA + C1 + SIN(ALFA1)
001165	DIFF =485(VL(K)-XO)
001170	1F(D1FF.LE.VL(K)/200.) GO TO 170
001175	XO = VL(K)
001177	50 TO 140
001177	170 VL(K) = VL(K)/12.
001202	OMEG'1 = SURT(DXOMEG = SUM2/3.)/FTA
001210	STRES(K)=E=ETA=C1=ALFA1/(XL(K)=12.)
001-14	
	C CONSTANT AMPLITUDE STRESS FOR EQUIV. FATIGUE DAMAGE
001220	GAMMAF = 1,843
001221	
	ARG = 1, + EXPM/2,
001224	IF(Expm, WE, 3.82) CALL GMMMA(ARG, GAMMAF, 1ER)
001231	STRESR = 1,414+(GAMMAF++(1,/EXPM)+STRES(K))
	c
001242	CALL FATIG(STRESR, SS(K), SEQ(K), CYC(K), LIFE(K), XMC(1), XMUP, RHO, XL
	1(K), 1P, XKS, S', EXPM, PI, OMEGA1)
	c
001265	180 CONTINUE
001271	PRINT 435, STATE(J), XMC(I)
001302	PRINT 425. (XL(K),STRES(K),SEQ(K),S5(K),CYC(K),LIFE(K),VL(K) ,K=1;
	1ºXL)
001337	190 CONTINUE
001344	200 CONTINUE
001344	RETURA
001345	END
100000000000-0	c

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SUBREUTINE COMUNISETA, ALFA1, OMEGA1, C1, XM, XMU, AP, XL, CO)

	The second
000014	3ET4 = x"/(x U=AP+XL+12.)
020020	ALF#: = 1.5775
000021	HETAT = "."
000022	CALL SECTATIASEC)
000024	4 DIFF = (BETA - BETAT)
000031	1F(4-5(31FF), LE.9ET4/203.) Gr TC 20
000036	ALFA1 = ALFA12 +CIFF
000040	PETAT = (C.S(ALFA1)/SIN(ALFA1))/4_FA1
000055	CALL SECONT(YSEC)
000057	1F(YSEC - SSEC .LE. 1.7) GO TO 4
000066	PRIVE 35,54 T4, ALFA1, OVEGA, C1, X4, X4U, AP, XL, CO, DIFF
050116	30 FORMAT(512,4,* ="ETA *,E12,4,* =ALFA1 *,E12,4,* =C*EGA *
	1 E12.4.0 =C1 0,E12.4.0 =X* 0,E12.4.0 =X*U 0 /
	2 ±12,4,0 = AP 0,E12,4,0 = XL 0,E12,4,0 = C3 0,
	3 F12,4,+ = = 1 (FF+)
000114	$20 \text{ CMEG} = 1 = 4 \text{ F}^1 = \text{ CC}/(\text{xL} \cdot 12 \cdot)$
000125	C1 = 1./ S.RT(X ++ (.5-SIN(2.+4LFA1)/(4.+ALFA1)+BETA+(SIN(4LFA1))++2
	1))
000157	PETUX
000160	ENP
	c

	SUBROUTINE FATIG(STRES, SS, SEG, CYC, LIFE, XMC, XMUP, RHO, XL, AP, XKS, SU, E 1XPY, P1,) EGA1)
000022	C HEAL LIFE
000022	TEN = XXC + (XMUP-RHO)/XMUP +(XMUP-RHO)+ XL + 12.44P
000030	SS = TE (AP
000031	1F(55,GT,S))50 TO 7000
000035	SEG = 573ES/(XKS = (1-SS/SU))
000040	IF(SEQ,LT.D)SEG=-SEQ
000043	CYC = (1102,3E3/5E0) ** EXPM
000050	LIFE = 2, •PI = CYC/ (OMEGA1 = 3600.)
000054	GO Tr 7:12
000055	700 CONTINUE
000055	SE3 #0
000056	CYC = 0
000056	
000057	PRINT 7011
000062	7301 FORMAT(64H IF SEG,CYC AND LIFE = 0 , STATIC STRESS IN STRING EXCEE
	1DS YIELD)
000062	7002 CONTINUE
000062	RETURN
000063	END

Construction Construction Construction C S. HBO TIVE GMMMA Construction Construction C P. RBORE Construction FOR A GIVEN ARGUMENT Construction C P. RBORE Construction Construction Construction Construction C Call GMMMA(XX,GX,IER) Construction Construction Construction C Call GMMA(XX,GX,IER) Construction Construction Construction Construction C Call GMMA(XX,GX,IER) Construction Construction Construction Construction C Call GMMA(XX,GX,IER) Construction Construction Construction Construction C						
C S. HRD TIVE GMMMA GMMM 3 C P. RRDSE GMMM 4 C P. RRDSE GMMM 4 C C.P.P. TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT GMMM 4 C C.P.P. TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT GMMM 4 C C.ALL GMMMA(XX,GX,IER) GMMM 11 DESCRIPTION OF PARAMETERS GMMM 12 C SX -THE RESULTANT GAMMA PUNCTION VALUE GMMM 13 C GX -THE RESULTANT GAMMA PUNCTION VALUE GMMM 14 IEF=-2EXILANT FOR CODE HURE GMMM 14 IEF=-2EXX ST 57, OVERFLOW, GX SET TO 1,0E75 GMMM 14 IEF=-2EXILANT FOR ODE HURE GMMM 12 C YETHON GMM 17 C IEF=-2EXX ST 57, OVERFLOW, GX SET TO 1,0E75 GMMM 14 C YETHON GMMM 23 GMMM 24 C YETHON GUNENZA GMMM 14 C YETHON GUNENZA GMMM 26 C YETHON GUNENZA GMMM 27 C YETHON GUNENZA GMMM 27 C				SUBROUTI E GYMMA(XX,GX,IER)	GMMM	330
C S. HRD TIVE GMMMA GMMM 3 C P. RRDSE GMMM 4 C P. RRDSE GMMM 4 C C.P.P. TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT GMMM 4 C C.P.P. TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT GMMM 4 C C.ALL GMMMA(XX,GX,IER) GMMM 11 DESCRIPTION OF PARAMETERS GMMM 12 C SX -THE RESULTANT GAMMA PUNCTION VALUE GMMM 13 C GX -THE RESULTANT GAMMA PUNCTION VALUE GMMM 14 IEF=-2EXILANT FOR CODE HURE GMMM 14 IEF=-2EXX ST 57, OVERFLOW, GX SET TO 1,0E75 GMMM 14 IEF=-2EXILANT FOR ODE HURE GMMM 12 C YETHON GMM 17 C IEF=-2EXX ST 57, OVERFLOW, GX SET TO 1,0E75 GMMM 14 C YETHON GMMM 23 GMMM 24 C YETHON GUNENZA GMMM 14 C YETHON GUNENZA GMMM 26 C YETHON GUNENZA GMMM 27 C YETHON GUNENZA GMMM 27 C		0			GMMM	10
C S. HBD TIVE GMMMA GMMM 4 C P. RBDSE GMMM 4 C C.P TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT GMMM 4 C GMM 6 GMMM 6 C CALL GMMMA(XX,GX,IER) GMMM 10 C GLALL GMMMA(XX,GX,IER) GMMM 10 C GLALL GMMMA(XX,GX,IER) GMMM 10 C GLALL GMMA(XX,GX,IER) GMMM 11 DESCRIPTION OF PARAMETERS GMMM 12 GX = THE RESULTANT GRAMA FUNCTION VALUE GMMM 14 IERAL XX IS - ITHIX, 000001 OF BEING A NEGATIVE INTEGER GMMM 14 IERMA 28 C IERAL XX IS - ITHIX, 000001 OF BEING A NEGATIVE INTEGER GMM 14 C IERAL XX IS - ITHIX, 000001 OF BEING A NEGATIVE INTEGER GMM 28 C SIRPO TINES AND FUNCTION SUBPROGRAMS PEQUIRED GMMM 28 C SIRPO TINES AND FUNCTION SUBPROGRAMS PEQUIRED GMMM 28 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMM 28 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMM 28 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMM 28 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMM 28 <		ē			GMMM	20
C S.HRD TIVE GONMA C P.RDSE C P.RDSE C P.TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT C P.RDSE C C P.TES THE GAMMA FUNCTION FOR A GIVEN ARGUMENT C C C C C C C C C C C C C C C C C C C		č				30
C P.RP3SE GMMM 8 C C.P. TES THE GAYMA FUNCTION FOR A GIVEN ARGUMENT GMMM 8 C C.G.C. P.TES THE GAYMA FUNCTION FOR A GIVEN ARGUMENT GMMM 8 C C.G.L.C.MMMA(XX,GX,IER) GMMM 10 C C.G.L.C.MMMA(XX,GX,IER) GMMM 10 C C.G.L.C.MMMA(XX,GX,IER) GMMM 12 C S.C.M.P.F.G.G.M.P.L.C.TION GMMM 12 C XX -THE AAGMENT FOR THE GAMMA FUNCTION VALUE GMMM 12 C XX -THE AAGGMENT FOR THE GAMMA FUNCTION VALUE GMMM 12 C XX -THE AAGGMENT FOR THE GAMMA FUNCTION VALUE GMMM 14 C IER=2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 12 C IER=2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 14 C NC*E GMMM 22 GMMM 22 C SI:RRO TIVES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 24 C YETHO' GMMM 24		č		E JOO TINE CHIMA		40
C P.RP38E C C.P.TES THE GAYMA FUNCTION FOR A GIVEN ARGUMENT C C CALL GMMM A (XX,GX,IER) C C CALL GMMMA (XX,GX,IER) C C C C CALL GMMMA (XX,GX,IER) C C C C C C C C C C C C C C C C C C C		č		Stight I the division		50
C C C P TES THE GAVMA FUNCTION FOR A GIVEN ARGUMENT GAMM A C C C GALL GAMMA (XX,GX,IER) C G C C C C C C C C C C C C C C C C C C				B. 20 AC-		0.000
C UCATE GMMM A C CALL GMMMA(XX,GX,IER) GMMM 10 C DESCRIPTION OF PARAMETERS GMMM 11 C XX - THE REGULTANT GAMMA FUNCTION VALUE GMMM 12 G XX - THE REGULTANT FROM CODE WERE GMMM 12 C IERAI VIEWERE GMMM 12 C SIRG TIVES AND FUNCTION SUBPROGRAMS REGUIRED GMMM 22 C NC'E GMMM 22 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 24 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 24 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 26 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 26 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 27						
C UCADE C CALL GMMMA(XX,GX,IER) C CALL GMMMA(X,GX,IER) C CALL GMMA(X,GX,IER) C CALL GMMMA(X,GX,IER) C CALL GMMA(X,GX,IER) C CALL GMMA(X,GX,IER)				CO P TES THE GATTA FUNCTION FOR A GIVEN ARGUMENT		0.00
C CALL GMMM A(XX,GX,1ER) C CALL GMMM A(XX,GX,1ER) C C CALL GMMM A(XX,GX,1ER) C DESCRIPTION OF PARAMETERS C X = THE ARGUMENT FOR THE GAMMA FUNCTION G X = THE RESULTANT GAMMA FUNCTION VALUE G X = THE RESULTANT FAROR CODE WERE C I TEMENT X IS #ITHIN,000001 OF BEING A NEGATIVE INTEGER GMMM 15 C TEMENT XX IS #ITHIN,000001 OF BEING A NEGATIVE INTEGER GMMM 17 C TEMENT XX IS #ITHIN,000001 OF BEING A NEGATIVE INTEGER GMMM 18 C S STREAT TO NO ERROR C S STREAT TO NO ERROR C S STREAT TO FUNCTION SUBPROGRAMS REQUIRED C NONE C NONE C NONE C S STREAT THE SAME FUNCTION SUBPROGRAMS REQUIRED C MMM 22 C NONE C METHON C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C MMM 23 C METHON C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C MMM 24 C MMM 25 C MMM 25 C MMM 25 C MMM 26 C MMM 26 C THON C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C MMM 26 C MMM 26 C MMM 27 C MMM 27 C MMM 27 C MMM 28 C MMM 29 C MMM 29 C MMM 29 C MMM 29 C MMM 29 C MMM 20 C MMM 29 C MMM 20 C MM 20 C MMM 20 C MMM 20 C MM 20						
C DESCRIPTION OF PARAMETERS GMMM 12 C XX -THE ARGUMENT FOR THE GAMMA FUNCTION GMMM 12 C XX -THE ARGUMENT FOR THE GAMMA FUNCTION GMMM 12 C GX -THE RESULTANT GAMMA FUNCTION GMMM 14 C IEFA-ESULTANT FAROR CODE WHERE GMMM 15 C IEFA-2 XX GT 57, OVERFLOW, GX SET TO 1.0E75 GMMM 16 C IEFA-2 XX GT 57, OVERFLOW, GX SET TO 1.0E75 GMMM 16 C NC*E GMMM 21 GMMM 26 C NC*E GMMM 26 GMMM 26 C NC*E GMMM 26 GMMM 27 C NC*E GMMM 26 GMMM 26 C NC*E GMMM 27 GMMM 27 C NC*E GMMM 28 GMMM 28 C NC*E GMMM 38 GMMM						
C DESCRIPTION OF PARAMETERS C MMM 13 C XX -THE RESULTANT GRAMMA FUNCTION VALUE GMMM 13 C XX -THE RESULTANT GRAMMA FUNCTION VALUE GMMM 14 C IERRO NO ERROR C IERCONSION RELATION SUBPROGRAMS REQUIRED C MMM 22 C NOCE C METHO C IERCONSION RELATION AND POLVNOMIAL APPROXIMATION C MMM 24 C IERCONSION RELATION AND POLVNOMIAL APPROXIMATION C MMM 26 C MMM 26 C IERCONSION RELATION AND POLVNOMIAL APPROXIMATION C MMM 26 C MMM 26 C MMM 27 C IERCONSION RELATION AND POLVNOMIAL APPROXIMATION C MMM 28 C MMM 29 C MMM 29 C MMM 29 C MMM 29 C MMM 29 C MMM 20 C MMM 20 C MMM 20 C IERCONSION RELATION INFORMATION C MMM 29 C MMM 20 C IERCONSION RELATION INFORMATION C MMM 29 C MMM 29				CALL GMMMA(XX,GX,IER)		
C XX -THE ARGUMENT FOR THE GAMMA FUNCTION GMMM 13 C GX -THE RESULTANT GAMA FUNCTION VALUE GMMM 14 C IER=2 SULTANT FAROR CODE WHERE GMMM 15 C IER=1 XX IS +ITH12,000001 OF BEING A NEGATIVE INTEGER GMMM 17 C IER=2 XX GT 57, OVERFLOW, GX SET TO 1.0E75 GMMM 16 C PEMARKS GMMM 21 C SIRPOTIVES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 22 C SIRPOTIVES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 23 C MC E GMMM 22 C YETMD" GMMM 24 C YETMD" GMMM 26 C PRIVEROY UNIVERSITY PRESS, 1955 GMMM 26 C PRIVETO' UNIVERSITY PRESS, 1955 GMMM 26 C MMM 26 C MMM 26 C MMM 27 C FIXES GMMM 27 C FIXES GMMM 27 C FIXES GMMM 28 C MMM 28 C MMM 28 C MMM 29 C MMM						
C GX -THE RESULTANT GAMMA PUNCTION VALUE GMMM 15 IER=20 NO ERROR GMMM 15 IER=1 XX IS #ITHIY,000001 OF BEING A NEGATIVE INTEGER GMMM 17 IER=2 XX GT 57,0VEMFLOW, GX SET TO 1.0E75 GMMM 17 IER=2 XX GT 57,0VEMFLOW, GX SET TO 1.0E75 GMMM 17 C NCE GMMM 20 C S: R40 TINES AND FUNCTION SUBPROGRAMS REGUIRED GMMM 26 C S: R40 TINES AND FUNCTION SUBPROGRAMS REGUIRED GMMM 26 C YETHO: GMMM 27 C MMM 26 C YETHO: GMMM 27 C YETHO:		C				120
C IET4ESULTANT FREOR CODE WERE GMMM 16 IET=0 NO CERROR GMMM 16 IET=1 XX IS #ITHI: 1000001 OF BEING A NEGATIVE INTEGER GMMM 17 IET=2 XX GT 57, OVERFLOW, GX SET TO 1.0E75 GMMM 18 C REMARKS GMMM 18 C REMARKS GMMM 21 GMMM 25 C SIRPO TINES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 25 C NONE GMMM 25				XX -THE ARGUMENT FOR THE GAMMA FUNCTION		
C IERED NO ERROR IERE1 XX IS +ITHI: ,000001 OF BEING & NEGATIVE INTEGER IERE2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 18 C IERE2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 20 GMMM 20 G		C		GX -THE RESULTANT GAMMA PUNCTION VALUE	GMMM	140
C IER=1 XX IS #ITHIN 1000001 OF BEING A NEGATIVE INTEGER GMMH 17 IER=2 XX GT 57, OVERFLOW, GX SET TO 1.0E75 GMMM 10 GMMM 10 C PEMARKS C NOTE C SIBO TIVES AND FUNCTION SUBPROGRAMS REGUIRED C NOTE C SIBO TIVES AND FUNCTION SUBPROGRAMS REGUIRED C NOTE C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C MMM 20 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C MMM 20 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C THE RETURN C THE RETURN		C		155-RESULTANT ERROR CODE WHERE		
C IER=2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 18 GMMM 12 GMMM 20 GMMM 20 GMM		C		ER=D NO ERROR		
C IER=2 XX GT 57, OVERFLOW, GX SET TO 1,0E75 GMMM 18 GMMM 12 GMMM 20 GMMM 20 GMM		C		IERE1 XX IS WITHIN .000001 OF BEING & NEGATIVE INTEGER	GMMM	170
C FEMARKS GMMM 10 C NC*E GMMM 21 C SIRPO TIVES AND FUNCTION SUBPROGRAMS REGUIRED GMMM 22 C NC*E GMMM 24 C NC*E GMMM 26 C T=FECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 26 C PRINCETON UNIVERSITY PRESS, 1955 GMMM 36 C GMMM 32 GMMM 36 C FETUSN GMMM 37 C GMMM 32 GMMM 36 C FETUSN GMMM 37 C GMMM 32 GMMM 36 C FETUSN GMMM 36 C GMMM 37 GMMM 37 C GMMM 37 GMMM 37 C		C				
C REMARKS GMMM 20 C YO'E GMMM 20 C SIRPO TIVES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 20 C NO'E GMMM 20 C METHON C METHON C BY C.HASTI'GS.JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 20 C PRIVCETO' UNIVERSITY PRESS, 1955 GMMM 30 C C FICKX-57.J4.6.4 GMMM 30 C C MMM 32 C MMM 3					GMMM	190
C '0'Ê GMMM 21 GMMM 22 C S:APO TIVES AND FUNCTION SUBPROGRAMS REQUIRED GMMM 23 C NC%E GMMM 24 C NC%E GMMM 25 C METHO: C MAN 20 C MMM 20 C		č		DE VADES		
C S:BRG_TIVES AVD_FUNCTION_SUBPROGRAMS_REQUIRED GMMM 23 C NC*E GMMM 24 C METHO: GMMM 26 C THE RECURSION_RELATION_AND_POLYNOMIAL_APPROXIMATION GMMM 26 C THE RECURSION_RELATION_AND_POLYNOMIAL_APPROXIMATION GMMM 26 C THE RECURSION_RELATION_AND_POLYNOMIAL_APPROXIMATION GMMM 26 C PRINCETO'_UNIVERSITY_PRESS, 1955 GMMM 26 C IF(X*-57,)*,f,f,4 GMMM 32 C000006 IF(X*-57,)*,f,f,4 GMMM 37 C00001 FERSE GMMM 32 C00001 IF(X*-57,)*,f,f,4 GMMM 37 C00001 FERSE GMMM 32 C00001 FERSE GMMM 32 C00011 FERSE GMMM 32 C00012 RETUFN GMMM 32 C00013 GX=1,2 GMMM 32 C00014 ERRE1,0E=6 GMMM 40 C00025 ID F(X=2,0150,50,15 GMMM 40 C00027 GC SEE IF X 1S NEAR NEGATIVE INTEGER OR ZERO GMMM 46 C00031 GC SEE I						
C SIRRO TIVES AND FUNCTION SUBPROGRAMS REGUIRED GMMM 23 C VC'E GMMM 24 C METHO: GMMM 25 C METHO: GMMM 25 C METHO: GMMM 26 C METHO: GMMM 27 C BY C.HASTI'GS.JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 28 C PRIVETO' UNIVERSITY PRESS, 1955 GMMM 28 C GMMM 20 C GMMM 20 C GMMM 27 C BY C.HASTI'GS.JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 28 C PRIVETO' UNIVERSITY PRESS, 1955 GMMM 30 C GMMM 30 C GMMM 30 C GMMM 30 C GMMM 34 C GMMM 34 C GMMM 34 C GMMM 35 C GMMM 40 C GMM 40 C GMM 40 C GMMM 40 C GMMM 40 C GMMM 40 C GMM		2		and the second		
C NC*E GMMM 24 C METHON GMMM 25 C T=F RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 27 BY C.HASTINGS, JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 27 GMMM 26 C PRIVCETON UNIVERSITY PRESS, 1955 GMMM 30 C IF(Xx=57.)%, 6, 4 GMMM 34 000010 4 IER=2 GMMM 34 000011 GXx=1.275 GMMM 34 000012 RETUTN GMMM 34 000013 6 X=XX 000014 ERR=1, 3E=6 GMMM 30 000015 IF(X=2,0)50,50,15 GMMM 40 000016 GX=1.3 GMMM 40 000017 IF(X=2,0)10,10,15 GMMM 43 000022 ID IF(X=2,0)110,110,15 GMMM 43 000021 GT F(X=1,0)60,120,110 GMMM 45 C SEE IF X 15 NEAR NEGATIVE INTEGER OR ZERO GMMM 45 000034 6C IF(X=EHR)62,62,80 GMMM 52 000034 6C IF(X=EHR)62,62,80 GMMM 52 0000034 6C IF(X=EHR)62,62,80		č		S SSS THESE WE SHOTLD SUBDACTURE BEAUTRES		
C METHO: GMMM 25 C METHO: GMMM 25 C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION GMMM 26 C BY C.HASTINGS.JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 26 PRINCETO: UNIVERSITY PRESS, 1955 GMMM 30 C IF(XX-57.)4.6.4 GMMM 31 C 000010 4 IER=2 C 000010 4 IER=2 C 000012 RETURN 000012 RETURN 000012 RETURN 000014 ERR=1.0E-6 C GMMM 35 000015 IER=1 000016 GX=1.2 C 000016 GX=1.2 C 000022 10 IF(X-2.0)57.50.15 C GMMM 40 000025 15 X=XX-1.7 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C GMMM 40 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C GMMM 40 C C GMMM 40 C C C GMMM 40 C C C GMMM 40 C C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C C GMMM 40 C C C GMMM 40 C C C C C GMMM 40 C C C C C C GMMM 40 C C C C C C GMMM 40 C C C C C C C C C C C C C C C C C C C		-				
C YETHO: THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION C BY CHASTINGS, JR, : IAPPROXIMATIONS FOR DIGITAL COMPUTERS:, PRINCETO' UNIVERSITY PRESS, 1955 C PRINCETO' UNIVERSITY PRESS, 1955 C PRINCETO' UNIVERSITY PRESS, 1955 C C C C C C C C C C C C C C C C C C C		5		10° E		
C THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION BY C.HASTINGS.JR., LAPPROXIMATIONS FOR DIGITAL COMPUTERSI, C PRINCETO' UNIVERSITY PRESS, 1955 C IF(Xx-57.)4.6.4 C IF(Xx-57.)4.6.4 C IF(Xx-57.)4.6.4 C C C C C C C C C C C C C C C C C C C				1999 - 1997 (P		
C BY C.MASTI'GS.JR., IAPPROXIMATIONS FOR DIGITAL COMPUTERSI, GMMM 28 PRINCETO' UNIVERSITY PRESS, 1955 C GMMM 30 C IF(XX-57,)4,6,4 000010 4 IER=2 000010 7 KX=1.E75 000012 RETURN 000012 RETURN 000013 6 X=XX 000014 ERR*1,0E=6 000015 IER* 000015 IER* 000017 IF(X-2,0)50,50,15 000022 10 IF(X+2,0)10,110,15 000022 10 IF(X+2,0)10,110,15 000025 15 X=X-1,0 000025 15 X=X-1,0 000025 C IF(X=1,0)60,120,110 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C MMM 48 000037 62 Y=FLCAT(INT(X))-X 000034 61 IF(1,0-Y=ERR)130,130,64 000026 C X:0T VEAP A NEGATIVE INTEGER OR ZER0 C MMM 54 000037 C X:0T VEAP A NEGATIVE INTEGER OR ZER0 C MMM 54 000036 C X:0T VEAP A NEGATIVE INTEGER OR ZER0 C MMM 54 000037 C X:0T VEAP A NEGATIVE INTEGER OR ZER0 C MMM 54 C MMM		c				
C PRINCETON UNIVERSITY PRESS, 1955 C PRINCETON UNIVERSITY PRESS, 1955 C GMMM 20 GMMM 31 C GMMM 32 C GMMM 42 C GMMM 43 C GMMM				THE RECURSION RELATION AND POLYNOMIAL APPROXIMATION		
C GMMM 30 C IF(Xx-57,)4,6,4 C GMMM 34 000010 4 IER=2 C GMMM 34 000012 RETUEN C GMMM 34 000012 RETUEN C GMMM 34 000012 RETUEN C GMMM 37 000013 6 x=xx C GMMM 37 000014 ERR1,0E=6 000015 IER=1 000016 GX=1.3 000017 IF(X=2,0)50,50,15 000017 IF(X=2,0)50,50,15 000028 ID IF(X=2,0)10,110,15 000029 ID IF(X=1,0)60,120,110 C GMMM 48 000031 50 IF(X=1,0)60,120,110 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C GMMM 48 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 C GMMM 52 000034 60 IF(A=ER)120,130,130,70 C GMMM 52 000046 64 IF(1,0-Y=ER)130,130,70 C GMMM 54 <td< td=""><td></td><td></td><td></td><td></td><td></td><td></td></td<>						
C C C C C C C C C C C C C C C C C C C				PRINCETO' UNIVERSITY PRESS, 1955		
C GMMM 32 000006 IF(xx-57,)*,6,4 GMMM 32 000010 4 IER=2 GMMM 35 000011 GX=1.E75 GMMM 36 000012 RETUEN GMMM 36 000013 6 x=xx GMMM 36 000014 ERR=1,0E=6 GMMM 39 000015 IER=* GMMM 41 000017 IF(x=2,0)50,50,15 GMMM 42 000021 10 IF(x=2,c)10,10,15 GMMM 42 000022 10 IF(x=2,c)10,10,15 GMMM 44 000027 Gx=Cx=X GMMM 45 000030 GO Tf 13 GMMM 45 000031 50 IF(x=1, 2)60, 120, 110 GMMM 45 000031 50 IF(x=1, 2)60, 120, 110 GMMM 45 000031 50 IF(x=1, 2)60, 120, 120, 64 GMMM 45 000034 60 IF(x=ER)62, 62, 80 GMMM 50 000037 62 Y=ELCAT(INT(X))=X GMMM 51 000036 64 IF(1,0=Y=ER)130, 130, 70 GMMM 53 000046 64 IF(1,0=Y=ER)130, 130, 70 GMMM 54 000046 <td< td=""><td></td><td>C</td><td></td><td></td><td></td><td></td></td<>		C				
000006 iF(xx-57,)*,6,4 GMMM 34 000010 4 iER=2 GMMM 35 000011 GX=1,E75 GMMM 36 000012 RETUE* GMMM 37 000013 6 x=xx GMMM 37 000014 ERR=1,0E=6 GMMM 30 000015 IER=* GMMM 40 000017 IF(x-2,0)50,50,15 GMMM 40 000022 10 IF(x-2,0)10,110,15 GMMM 42 000025 15 x=x-1,0 GMMM 43 000027 Gx=Gx=X GMMM 43 000027 GX=GX=X GMMM 44 000030 GO Tf 13 GMMM 45 000031 50 IF(x=1,0)*67,120,110 GMMM 48 000031 50 IF(x=1,0)*67,120,110 GMMM 49 000031 50 IF(x=1,0)*67,120,110 GMMM 48 000034 6C IF(x=ERB)*62,62,80 GMMM 50 000037 62 *=1F x IS NEAR NEGATIVE INTEGER OR ZER0 GMMM 53 000046 64 IF(1,0-*=ERB)*130,130,70 GMMM 53 000046 64 IF(1,0-*=ERB)*130,130,70 GMMM 54						
000010 4 IER=2 GMMM 35 000011 GX=1.E75 GMMM 36 000012 RETUEN GMMM 37 000013 6 x=xx GMMM 36 000014 ERR=1.0E=6 GMMM 30 000015 IER=" GMMM 40 000016 GX=1.3 GMMM 41 000022 10 IF(X=2.0150.50.15 GMMM 42 000025 15 x=x=1.0 GMMM 43 000027 GX=(X=X) GMMM 43 000027 GX=(X=X) GMMM 44 000030 GO TC 13 GMMM 44 000031 50 IF(X=1.0160.120.110 GMMM 46 000031 50 GTC 13 GMMM 46 000031 50 IF(X=1.0160.120.110 GMMM 47 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZER0 GMMM 47 000034 60 IF(X=ERR)62.62.80 GMMM 50 000037 62 Y=FLCAT(INT(X))=X GMMM 51 000034 62 IF(A=S(Y)=ERPI130.130.64 GMMM 53 000046 64 IF(1.0=Y=ER)130.130.70 GMMM 54 000046 64 IF(1.0=Y=ER)130.130.70 GMMM 54 C X=0T=ERP A		C			GMMM	320
000010 4 IERe2 GMMM 35 000011 GX=1.E75 GMMM 36 000012 RETUFN GMMM 37 000013 6 X=XX GMMM 38 000014 ERRE1.0E=6 GMMM 38 000015 IERe* GMMM 40 000016 GX=1.3 GMMM 40 000027 IF [X+2.0]50.50.15 GMMM 41 000027 ID [F [X+2.0]10.110.15 GMMM 43 000027 GX=(X+X) GMMM 44 000027 GX=(X+X) GMMM 44 000030 GO Tf 13 GMMM 45 000031 50 IF (X=1.0)60.120.110 GMMM 46 0000331 50 IF (X=1.0)60.120.110 GMMM 46 000034 61 IF (X=ERR)62.62.80 GMMM 45 000037 62 Y=ELCAT(INT(X))-X GMMM 50 000046 64 IF (1.0-Y=ERR)130.130.64 GMMM 53 000046<	000006			[F(XX=57.)4,6,4	GMMM	340
000011 Ax=1.E75 GMMM 36 000012 RETUEN GMMM 37 000013 6 x=xx GMMM 38 000014 ERR=1.JE=6 GMMM 39 000015 IER=* GMMM 41 000017 IF(x=2,0)50,50,15 GMMM 42 000022 10 IF(x=2,0)50,50,15 GMMM 42 000027 Sx=Cx=X GMMM 43 000027 GX=GX=X GMMM 43 000027 GX=Cx=X GMMM 44 000027 GX=CX=X GMMM 45 000030 GO Tf 13 GMMM 45 000031 50 IF(x=1,2)67,120,110 GMMM 46 000031 50 IF(x=1,2)67,120,110 GMMM 47 C SEE 1F X 15 NEAR NEGATIVE INTEGER OR ZERO GMMM 48 000034 60 IF(x=1,2)62,62,80 GMMM 50 000034 62 IF(A=5(Y)=ER=1)30,130,64 GMMM 53 000046 64 IF(1,0=Y=ER=1)30,130,70 GMMM 54 000046 64 IF(1,0=Y=ER=1)30,130,70 GMMM 54	000010		4		GMMM	350
000012 RETUR GMMM 37 000013 6 x=xx GMMM 37 000014 ERR1,0E=6 GMMM 39 000015 IER* GMMM 30 000016 Gx=1,0 GMMM 40 000017 IF(x=2,0)50,50,15 GMMM 40 000022 10 IF(x=2,0)10,110,15 GMMM 43 000025 15 x=x-1,0 GMMM 43 000027 Gx=Gx=X GMMM 43 000030 GO Tf 13 GMMM 43 000031 50 IF(x=1,0)60,120,110 GMMM 43 000034 60 IF(x=ERR)62,62,80 GMMM 43 000034 62 IF(A=ERR)62,62,80 GMMM 53 000034 62 IF(A=S(Y)=ERPR)130,130,64 GMMM 53 000046 64 IF(1,0=Y=ERR)130,130,70 GMMM 53 000046 64 IF(1,0=Y=ERR)130,130,70 GMMM 54 000046 C X=0T=ERP A NEGATIVE INTEGER 07 ZER0					GMMM	360
000013 6 x=xx GMMM 3a 000014 ERR#1,0E=6 GMMM 30 000015 IER#1,0E=6 GMMM 40 000016 GX=1,3 GMMM 40 000017 IF(X=2,0)50,50,15 GMMM 41 000027 IS X=X-1,0 GMMM 43 000027 GX=(X=X) GMMM 43 000027 GX=(X=X) GMMM 44 000030 GO T(1) GMMM 44 000031 50 IF(X=1,0)60,120,110 GMMM 44 000031 50 IF(X=1,0)60,120,110 GMMM 46 000031 50 IF(X=1,0)60,120,110 GMMM 47 000034 60 IF(X=1,0)60,120,110 GMMM 46 000037 52 SEE IF X 1S NEAR NEGATIVE INTEGER OR ZER0 GMMM 51 000037 62 Y=FLCAT(INT(X))-X GMMM 51 000042 IF(A=S(Y)=ER+130,130,130,70 GMMM 53 000046 64 IF(1,0=Y=ER+130,130,70 GMMM 54 000046 C X=000 SER GMMM 54 000046 C <td< td=""><td></td><td></td><td></td><td></td><td>GMMM</td><td>370</td></td<>					GMMM	370
D00014 ERR:1,0E=6 GMMM 39 D00015 IER* GMMM 40 D00017 IF(X=2,0)50,50.15 GMMM 42 D00022 10 IF(X=2,0)10,10,15 GMMM 42 D00025 15 x=x-1,0 GMMM 44 000027 GX=GX=X GMMM 44 000027 GX=GX=X GMMM 44 000021 50 IF(X=1,0)60,120,110 GMMM 44 000031 50 IF(X=1,0)60,120,110 GMMM 46 000031 50 IF(X=1,0)60,120,110 GMMM 48 000031 50 IF(X=1,0)60,120,110 GMMM 48 000034 60 IF(X=ERR)62,62,80 GMMM 50 000034 60 IF(X=ERR)62,62,80 GMMM 51 000037 52 Y=FLCAT(INT(X))-X GMMM 52 000046 64 IF(1,0-Y=ERR)130,130,64 GMMM 53 000046 64 IF(1,0-Y=ERR)130,130,70 GMMM 54 C X-0T -ERP 1A NEGATIVE INTEGER 0R ZER0 GMMM 54			6		GMMM	380
000015 IER* GMMM 40 000016 GX1.3 GMMM 40 000017 IF(X=2,0)50,50,15 GMMM 40 000022 10 IF(X=2,0)10,110,15 GMMM 40 000025 15 X=X-1,0 GMMM 40 000027 GX=GX=X GMMM 40 000027 GX=GX=X GMMM 40 000030 GO Tf 13 GMMM 40 000031 50 IF(X=1,0)60,120,110 GMMM 40 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 40 000034 6C IF(X=ERR)62,62,80 GMMM 50 000037 62 Y=FLCAT(INT(X))=X GMMM 50 000046 64 IF(1,0=Y=ERR)130,130,64 GMMM 53 000046 64 IF(1,0=Y=ERR)130,130,70 GMMM 54 C X=0T=ERP A NEGATIVE INTEGER OR ZERO GMMM 54					GMMM	390
000016 GX1.3 GMMM 41 000017 IF(X+2,0)50,50,15 GMMM 42 000022 10 IF(X+2,0)10,110,15 GMMM 43 000025 15 x=x-1,0 GMMM 44 000027 GX=GX+X GMMM 44 000030 GO TC 10 GMMM 44 000031 50 IF(X+1,0)60,120,110 GMMM 46 000031 50 IF(X+1,0)60,120,110 GMMM 47 000034 60 IF(X+1,0)60,120,110 GMMM 46 000034 60 IF(X+1,0)60,120,110 GMMM 46 000034 60 IF(X+1,0)60,000,000 GMMM 46 000037 62 Y=FLCAT(INT(X))-X GMMM 50 000046 64 IF(1,0-Y-ERR)130,130,64 GMMM 53 000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 000047 C GMMM 54 000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 000046 C X-0T YEAR GMMM 56 000046 C					GMMM	400
000017 IF(X+2,0)50,50,15 GMMM 42 000022 10 IF(X+2,0)10,110,15 GMMM 43 000025 15 x=x-1,0 GMMM 43 000027 Gx=Gx=X GMMM 43 000030 GO T (1) GMMM 43 000031 50 IF(X=1,0)60,120,110 GMMM 48 000034 60 IF(X=1,0)60,120,110 GMMM 48 000034 60 IF(X=ERR)62,62,80 GMMM 50 000034 60 IF(X=ERR)62,62,80 GMMM 50 000034 62 Y=ELCAT(INT(X))-X GMMM 50 000046 64 IF(1,0-Y=ERR)130,130,64 GMMM 53 000046 64 IF(1,0-Y=ERR)130,130,70 GMMM 54 C X-OT -EAR A NEGATIVE INTEGER OR ZERO GMMM 54						
000022 10 IF(X=2,()110,110,15 GMMM 43 000025 15 X=X=1,0 GMMM 44 000027 GX=GX=X GMMM 44 000031 GO T(1) GMMM 45 000031 50 IF(X=1,0)60,120,110 GMMM 46 000034 6C IF(X=1,0)60,120,110 GMMM 47 000034 6C IF(X=HR)62,62,80 GMMM 45 000037 62 Y=FLCAT(INT(X))-X GMMM 52 000042 IF(A=S(Y)-ERP)130,130,64 GMMM 53 000046 64 IF(1,0=Y=ER)130,130,70 GMMM 54 C X=0T=ERP A NEGATIVE INTEGER OR ZERO GMMM 54						
000025 15 X=X-1,0 GMMM 44 000027 GX=GX=X GMMM 45 000030 GO TC 13 GMMM 45 000031 50 IF(X=1,0)60,120,110 GMMM 47 C SEE IF X 1S NEAR NEGATIVE INTEGER OR ZERO GMMM 46 C SEE IF X 15 NEAR NEGATIVE INTEGER OR ZERO GMMM 46 000034 6C IF(X=ERR)62,62,80 GMMM 50 000037 62 Y=FLCAT(INT(X))-X GMMM 51 000042 IF(AFS(Y)=ERR)130,130,64 GMMM 53 000046 64 IF(1,0=Y=ERR)130,130,70 GMMM 54 C X :0T YEAR A NEGATIVE INTEGER OR ZERO GMMM 54					1	
000027 GX 6 GX #X GMMM 45 000030 GO TC 13 GMMM 45 000031 50 IF(X=1,3)60,120,110 GMMM 46 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 46 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 47 000034 60 IF(X=ER3)62,62,80 GMMM 50 000034 62 IF(A=C(Y)=ER3)130,130,64 GMMM 51 000046 64 IF(1,0=Y=ER3)130,130,70 GMMM 54 C X :.0T 'EAP A NEGATIVE INTEGER OR ZERO GMMM 54						
000030 G0 T7 13 GMMM 46 000031 50 IF(X=1,0)60,120,110 GMMM 47 000034 50 IF(X=1,0)60,120,110 GMMM 47 000034 60 IF(X=1,0)60,120,110 GMMM 48 000034 60 IF(X=ERR)62,62,80 GMMM 50 000037 62 Y=FLCAT(INT(X))-X GMMM 52 000042 IF(A=S(Y)=ERP)130,130,64 GMMM 53 000046 64 IF(1,0=Y=ERP)130,130,70 GMMM 54 C X :0T -ERP A NEGATIVE INTEGER OR ZERO GMMM 54			12			
000031 50 IF(X=1,0)60,120,110 GMMM 47 C SEE IF X 1S NEAR NEGATIVE INTEGER OR ZERO GMMM 49 000034 60 IF(X=ERR)62,62,80 GMMM 50 000037 62 V=FLCAT(INT(X))-X GMMM 51 000037 12F(AFS(Y)-ERR)130,130,64 GMMM 53 000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 C X :0T -ERR)130,130,70 GMMM 54 C X :0T -ERR)130,130,70 GMMM 54						0.7.5
C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 48 C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 50 000034 6C IF(X=EHR)62,62,80 GMMM 50 000037 62 Y=ELCAT(INT(X))-X GMMM 51 000046 IF(AFS(Y)=ERF)130,130,64 GMMM 53 000046 64 IF(1,0-Y=ERR)130,130,70 GMMM 54 C X :QT Y=ER GMMM 54 C X :QT Y=EAR A NEGATIVE INTEGER OR ZERO			582			
C SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO GMMM 40 C GMMM 50 GMMM 50 D00034 6C IF(X=ER)62,62,80 GMMM 51 D00037 62 Y=FLCAT(INT(X))-X GMMM 52 D00042 IF(AFS(Y)-ER)130,130,64 GMMM 53 D00046 64 IF(1,0-Y=ER)130,130,70 GMMM 54 C X ::QT :=ER A NEGATIVE INTEGER OR ZERO GMMM 54 C X ::QT :=ER A NEGATIVE INTEGER OR ZERO GMMM 54	000031	10180	50	IF(X+1,3)60,120,110		
C GMMM 50 D0D034 6C IF(X=EHR)62,62,80 GMMM 51 D0D037 62 Y=FLCAT(INT(X))=X GMMM 52 D0D042 IF(AFS(Y)=ERR)130,130,64 GMMM 53 D00046 64 IF(1,0=Y=ERR)130,130,70 GMMM 54 C X ::0T :=ERP A NEGATIVE INTEGER OR ZERO GMMM 56						
000034 6C IF(X=ER)62,62,80 GMMM 51 000037 62 Y=ELCAT(INT(X))-X GMMM 52 000042 IF(AFS(Y)=ER)130,130,64 GMMM 53 000046 64 IF(1,0-Y=ER)130,130,70 GMMM 54 C X .CT .EAP A NEGATIVE INTEGER OR ZERO GMMM 54		¢		SEE IF X IS NEAR NEGATIVE INTEGER OR ZERO		
D00037 62 Y=FLCAT(INT(X))-X GMMM 52 000042 IF(ARS(Y)-ERH)130,130,64 GMMM 53 000046 64 IF(I.0-Y-ERH)130,130,70 GMMM 54 C X .0T .EAP A NEGATIVE INTEGER OR ZERO GMMM 54		C				
000042 IF(ARS(Y)-ERE)130,130,64 GMMM 53 000046 64 IF(1.0-Y-ERE)130,130,70 GMMM 54 C X .0T .EAP A NEGATIVE INTEGER OR ZERO GMMM 56	000034		60	IF(X=EHR)62,62,80		
000042 IF(ARS(Y)-ERH)130,130,64 GMMM 53 000046 64 IF(1.0-Y-ERR)130,130,70 GMMM 54 C X .CT .EAP A NEGATIVE INTEGER OR ZERO GMMM 56					GMMM	520
000046 64 IF(1,0-Y-ERR)130,130,70 GMMM 54 C GMMM 56 C X .CT .EAP A LEGATIVE INTEGER OR ZERO GMMM 56					GMMM	530
C X CT SEAP & NEGATIVE INTEGER OF ZERO GMMM 55			64		GMMM	540
C XCT -EAP A NEGATIVE INTEGER OF ZERO GMMM 56		с	200	はないための シリス いたいたち ままき ひまう 人名	GMMM	550
		č		X GT JEAR A NEGATIVE INTEGER OR TERO		
					GMMM	570
		27			120000	19581

000052	70	IF(X-1.2)47,40,110	GMMM	580
000055			GMMM	590
000056		X=X+1.	GMMM	600
000060		90 TS 73	GMMM	610
000060	110	Y=X-1,	GMMM	620
000062		GY=1.0+Y=(-0.5771017+Y=(+0.9858540+Y=(-0.8764218+Y=(+0.8328212+	GMMM	630
0.0000		1Y+(-1,5534729+Y+(+0.2548205+Y+(+0.05149930)))))))	GMMM	640
000100		GX=GX+GY	GMMM	650
000102	120	RETU	GMMM	660
 000103		TER#1	GMMM	670
000104	204	HETURY	GMMM	680
000105		FND	GMMM	690

EXAMPLE PHONEEV--ESTAGE = 5 FT. DD = 1 FT. XDOT = 30 FT/HR. H = 51.73 FT

NEN DRILL FIPE

PIPE DIA "ETCH. 1	5,00
CROSS-SICTICAL ASE , S'. I'.	5.66
VEIGHT PER-FOOT, L	19.50
TENSILE STAL STA. PII	160000.00
ELASTIC MODILE, PAI	30000000.00
CIAMETER OF RUPRES, 1	7.00

HEAVE RESPONSE TO SHELLS

 RUBHER SPACING, FT
 =
 5.00

 NG, OF DAAPING DISKS
 =
 1.00

 DIAMETER OF DAAPING DISKS, FT
 =
 1.00

 RATIO OF AAVE HEIGHT TO AAVE LENGTH
 =
 0.09

 FRACTION OF AAVE HEIGHT TO AAVE LENGTH
 =
 0.09

 WAVE HEIGHT, FT
 =
 1.00

 WAVE HEIGHT, FT
 =
 1.00

 WEIGHT AT END OF STAILG, LB
 0.00

 STRING LENGTH DY. STRESS
 EGUIV STRESS
 STATIC STRESS
 LIFE IN CYC
 LIFE IN HOURS
 REL DISP, FT
 ABS DISP, FT

 10000,00
 7747,33
 9628,44
 30008,55
 7,3182E+07
 4,8950E+04
 1.49
 1.24

 22000,00
 3726,85
 7049,82
 66018,80
 2,4074E+08
 3,5424E+05
 1.74
 1.49

 WAVE HEIGHT, FT
 =
 20,00
 0,00
 0.00
 0.00
 0.00

 STRING LENGTH
 DYN STRESS
 EQUIV STRESS
 STATIC STRESS
 LIFE IN CYC
 LIFE IN HOURS
 REL DISP, FT
 ABS DISP, FT

 10000.00
 20479,39
 28554,89
 30008.55
 1.1505E+06
 7.6995E+02
 4.43
 4.76

 22000.00
 14292,99
 27037.07
 66018.80
 1.4174E+06
 2.0857E+03
 6.67
 3.48

 WAVE HEIGHT, FT
 =
 10.00
 weight at END OF STHING, LB
 35000.00
 1.4174E+06
 1.4174E+03
 1.4174E+03

 STRING LLNGTH
 0Y: STRESS
 EQUIV STRESS
 STATIC STRESS
 LIFE IN CYC
 LIFE IN HOURS
 REL DISP, FT
 ABS DISP, FT

 10000.00
 7474.41
 10663.94
 35394.70
 4.9538E+07
 3.8865E+04
 1.81
 1.49

 2200G.00
 3597.57
 7219.01
 71404.95
 2.1989E+08
 3.4933E+05
 1.80
 1.76

 WAVE HEIGHT, FT
 =
 20.00
 WAVE HEIGHT, FT
 =
 20.00
 1.80
 1.76

STRING LENGT	A DY STRESS	EQUIV STRESS	STATIC STRESS	LIFE IN CYC	LIFE IN HOURS		ABS DISP. FT
10000.00	21439.88 14-53.13	29162.13 28199.48	35394,70	1.0616E+06 1.2069E+06	8,3290E+02 1,9173E+03	4.95	5,41 4,42

HEAVE RESPONSE TO RINDON SEAS

 RUBBER SPACING, FT
 =
 5.00

 NO. OF DAMPING DISKS
 =
 1.00

 DIAMETER OF DAMPING DIS-S, FT
 =
 1.00

 SEA STATE, FT SCD
 =
 25.23

 WEIGHT AT END OF STRING, LB
 0.00

STRING LENGTH	**S STRESS	EGUIV STRESS	STATIC STRESS	LIFE IN CYC	LIFE IN HOURS	RMS REL DISP. FT
10000.00	555,33	1260.29	30008.55	1.7290E+11	8.5239E+07	.12
22000.00	*56,12	2059.59	66018.80	2.6481E+10	1.2908E+07	.31
SEA STATE, FT SOD WEIGHT AT END OF STRIN	∎ ¢G.LB ■	82.58				

STRING LENGTH 10000,00 22000.00	R'S STRESS 1006,70 1188,58	EQUIV STRESS 2284.67 3730.98	STATIC STRESS 30008.55 66018.80	LIFE IN CYC 1.7819E+10 2.7367E+09	LIFE IN HOURS 8,7846€+06 1,3338E+06	RMS RE	. DISP, .21 .55	FT
SEA STATE, FT SOD WEIGHT AT END OF STRI		5,23						
STEING LENGTH 10000.20 22009.70	- S STRESS 485,54 537,45	EGUIV STRESS 1149.54 2122.61	STATIC STRESS 35394.70 71404.95	LIFE IN CYC 2.45696+11 2.36016+10	LIPE IN HOURS 1,2142E+08 1,1523E+07	RMS RE	. 015P, .12 .32	PT
SEA STATE, FT SOD WEIGHT AT END OF STAI		2.58 0.00						

STRING LENGTH	R'S STRESS	EQUIV STRESS	STATIC STRESS	LIFE IN CYC	LIFE IN HOURS	RMS REL DISP, PT
10000,00	180.03	2083.51	35394.70	2.55398+10	1.25228+07	.21
22000,00	1154.62	3844.73	71404.95	2.44008+09	1.19128+06	.96

59

ROLL AND PITCH - RIGIN JUDF HEIGTH OF GLIDE, FT = 51,73 GUIDE DESIGNED FOR R.ST DEGREES ROLL AND FITCH - FLEXILLE GLIDE HEIGTH OF GUIDE, FT = 51,73 GUIDE DESIGNE, FOR 1000,000 FT STRING

RCTARY DRILLING RATE, RET = 60,00 LINEAR DRILLING RATE, FTAR = 30,00 RUBBER SPACING, FT = 5.00

TOTAL WEIGHT AT END OF STRING, LH = 0.00 FLEXIBLE GUINE DESIGN PARIMETERS DIST, FROM TIP 01TSIDE DIAMETER MOMENT OF INERTIA 62.08 9.44 44.02 124.15 0.40 127.30 176.23 10.08 249.86 246.30 10.80 411.69 310.38 11.53 612.76 372.46 12.26 A95.15 434.53 12.97 1132.79 496.61 13.66 1451.69 555.68 14.32 1309.87 620.76 14.97 2207.32 MAX STRESS IN GUIDE FOR A GULAR DEVIATION OF 8.50 DEGREES = 50518.71 PSI

ANGULAR DEVIATION, DEG

= 5,00

STRING LEAGTH BENNING STRESS EQUIV, STRESS STATIC STRESS LIFE IN CYCLES LIFE IN HOURS NUMBER OF HOLES

RIGID GUIDE 10000.00	19*18.43	27103,91	30008,55	1,4041E+06	3,9003E+02	601.4
FLEXIBLE GUIDE	10501.05	14771,64	30008.55	1.4268E+07	3,9654E+03	8455.1
RIGID GUIDE 22000.00	22063.09	41735,23	66018,80	2,6994E+05	7,4983E+01	115,6
FLEXIBLE GUIDE 22000,00	12333,37	23330,18	66018.80	2.4896E+06	6,9157E+02	1517.6

ANGULAR DEVIATION, CEG = 8,50

STRING LENGTH BENDING STRESS EQUIV, STRESS STATIC STRESS LIFE IN CYCLES LIFE IN HOURS NUMBER OF HOLES RIGID GUIDE 353,8 19813.43 1.4041E+06 3,90032+02 10000.00 27103,91 30005.55 FLEXIBLE GUIDE 1.8795E+06 30008.55 5.2210E+02 1113.8 10000.00 18361,78 25111.78

RIGID GUIDE 22000.00	22063.09	41735,23	66018.80	2.6994E+05	7.49832+01	68.0
FLEXIBLE GUIDE 22000.00	20964.72	39661.31	66018.80	3.2796E+05	9.1100E+01	199,9

TOTAL WEIGHT AT END OF STRI'G. LB = 35000.00

FLEXIBLE GUIDE DESIGN	PARAMETERS		
DIST. FROM TIP	OUTSIDE DIAMETER MO	MENT OF INERTIA	
62.08	8,87	47,16	
124,15	9,46	137,51	
186.23	10.18	271.04	
248,30	10.94	447.76	
310.38	11,71	667.66	
372,46	12,47	930,75	
434,53	13,21	1237.03	
496.61	13,92	1586,49	
558.68	14.61	1979.14	
620.76	15,27	2414.97	
	A ANGULAR DEVIATION OF		51691.45 PS!

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60

- 23	5.00				
BENTI & STRESS	EQUIV, STRESS	STATIC STREES	LIFE IN CYCLES	LIFE IN HOURS	NUMBER OF HOLES
20161,36	28764.76	35394,70	1.1188E+06	3.1077E+02	479.2
11055.29	15772,89	35394.70	1.1106E+07	3,0850E+03	6606.0
22389,45	44927,43	71404,95	2.0370E+05	5,65835+01	87,3
12541,45	25166.09	71404.95	1.86412+06	5.1781E+02	1155.4
DEG .	8.50				
BENDING STRESS	EQUIV. STRESS	STATIC STRESS	LIFE IN CYCLES	LIFE IN HOURS	NUMBER OF HOLES
20161,36	28764,76	35394.70	1,11885+06	3.1077E+02	281.9
18794.00	26813.91	35394.70	1,4630E+06	4,06396+02	870.2
22389.45	44927,43	71404.95	2.03702+05	5,65838+01	51.3
21320,47	42782,36	71404.95	2,45568+05	6.8211E+01	149.6
ARDS READ , 7300, CM70000, 442					
7 SEC. 7 SEC.					
	BENNI 16 STRESS 20161,36 11055,29 22389,45 12541,45 DEG BENDING STRESS 20161,36 18794.00 22389.45 21320,47 FR KNOWLEDGE 3,2 4RDS READ ,7300,CM70000,442	BENNI & STRESS EQUIV. STRESS 20161.36 28764.76 11055.29 15772.89 22389.45 44927.43 12541.45 25166.09 DEG 8.50 BENDING STRESS EQUIV. STRESS 20161.36 28764.76 18794.00 26813.91 22389.45 44927.43 21320.47 42782.36 F KNOWLEDGE 3.2 1739 07/10/70 ARDS READ .7300.CM70000.442102280701 HGP .94 SWRI TERMINAL ***** .54670C) 7 SEC. 2 SEC.	BENNI & STRESS EQUIV. STRESS STATIC STRESS 20161.36 28764.76 35394.70 11055.29 15772.89 35394.70 22389.45 44927.43 71404.95 12541.45 25166.09 71404.95 DEG 8.50 8.50 BENDING STRESS EQUIV. STRESS STATIC STRESS 20161.36 28764.76 35394.70 18794.00 26813.91 35394.70 22389.45 44927.43 71404.95 21320.47 42782.36 71404.95 21320.47 42782.36 71404.95 24389.45 44927.43 71404.95 258.0.47 42782.36 71404.95 21320.47 42782.36 71404.95 24300.000.442102280701 HGP 1404.95 194 SWRI TERMINAL ***** .5457000 17 SEC. 7 5 2 SEC. 5 5	BENNI 16 STRESS EQUIV. STRESS STATIC STRESS LIFE IN CYCLES 20161.36 28764.76 35394.70 1.1188E+06 11055.29 15772.89 35394.70 1.1106E+07 22389.45 44927.43 71404.95 2.0370E+05 12541.45 25166.09 71404.95 1.8641E+06 DEG 8.50 BENDING STRESS EQUIV. STRESS STATIC STRESS LIFE IN CYCLES 20161.36 28764.76 35394.70 1.1188E+06 18794.30 26813.91 35394.70 1.4630E+06 18794.30 26813.91 35394.70 1.4630E+06 2389.45 44927.43 71404.95 2.4556E+05 21320.47 42782.36 71404.95 2.4556E+05 FF KNOWLEDGE 3.2 1739 07/10/70 1.4630E+06 1.4556E+05 FF KNOWLEDGE 3.2 1739 07/10/70 1.4630E+05 1.4556E+05 FF KNOWLEDGE 3.2 1739 07/10/70 1.4556E+05 1.404.95 2.4556E+05 FF KNOWLEDGE 3.2 1739 07/10/70 1.750E, C.1 1.5556E+05 1.5670, C.1 75 SEC. 58EC. 58EC. 58EC	BENNI & STRESS EQUIV. STRESS STATIC STRESS LIFE IN CYCLES LIFE IN HOURS 20161.36 28764.76 35394.70 1.1188E+06 3.1077E+02 11355.29 15772.89 35394.70 1.1106E+07 3.0850E+03 22389.45 44927.43 71404.95 2.0370E+05 5.4583E+01 12541.45 25186.09 71404.95 1.8641E+06 5.1781E+02 DEG 8.50

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APPENDIX

DERIVATION OF HEAVE DYNAMICS MODAL SOLUTION

In this Appendix, the modal solution for the heave response, Eqs. (2.4) to (2.17), is derived. The governing equations and boundary conditions are:

$$\rho A \frac{\partial^2 u}{\partial t^2} = A E \frac{\partial^2 u}{\partial x^2} - c \frac{\partial u}{\partial t}$$
(A.1)

where

$$\rho A = \mu A + M\delta(x - \ell) \tag{A.2a}$$

$$c = D + C\delta(x - \ell) \tag{A.2b}$$

and

$$u(0,t) = W(t) \tag{A.3a}$$

$$\frac{\partial u}{\partial x}\Big|_{x=\varrho} = -\frac{M}{AE} \frac{\partial^2 u}{\partial t^2}\Big|_{x=\varrho}$$
(A.3b)

The beginning point is the solution to the free vibration problem, $\overline{u}(x, t)$, wherein W(t) = 0. This is done by assuming a separable solution of the form

$$\overline{u}(x,t) = X(x)T(t) \tag{A.4}$$

whereupon Eq. (A.1) becomes

$$\frac{\ddot{T}}{T} + \frac{D}{\mu A} \frac{\dot{T}}{T} = \frac{E}{\mu} \frac{X''}{X} = \text{constant} = -\frac{E\alpha_1^2}{\mu \ell^2} = -\frac{c_0^2 \alpha_1^2}{\ell^2}$$
(A.5)

In this expression, the superscripted primes and dots indicate, respectively, differentiation with respect to x and t, and the form of the constant has been chosen for later convenience.

The mode functions are seen to be governed by

$$X'' + \left(\frac{\alpha_1}{\varrho}\right)^2 X = 0 \tag{A.6}$$

The first mode solution to Eq. (A.6) is

$$X(x) = a_1 \sin \frac{\alpha_1 x}{\varrho} + a_2 \cos \frac{\alpha_1 x}{\varrho}$$
(A.7)

The constant a_2 vanishes by virtue of (A.3a). In satisfying boundary condition (A.3b), one is led to

$$X'(\mathfrak{Q})T(t) = -\frac{M}{AE}X(\mathfrak{Q})\ddot{T}(t)$$
(A.8)

If the system is only *lightly damped*, the unknown time function is given approximately by solutions to

$$\ddot{T} + \left(\frac{c_0 \alpha_1}{\varrho}\right)^2 T = 0 \tag{A.9}$$

The first mode solution to this equation is

$$T(t) = b_1 \sin \frac{c_0 \alpha_1 t}{\varrho} + b_2 \cos \frac{c_0 \alpha_1 t}{\varrho}$$
(A.10)

Substituting (A.10) into (A.8), one arrives at the frequency equation

$$\operatorname{ctn} \alpha_1 = \beta \alpha_1 \tag{A.11}$$

which defines α_1 implicitly. In this expression, $\beta = M/\mu A \ell$.

Equation (A.11) is trancendental in α_1 and defines an infinite set of α_1 's. For the first-mode solution of interest here, one is concerned only in the fundamental value of α_1 . The first-mode solution can thus be expressed as

$$\overline{u}(x,t) = a_1 T(t) \sin \frac{\alpha_1 x}{\ell}$$
(A.12)

The forced-vibration response can be constructed from the normal mode solution using the method of virtual work, as outlined by Timoshenko(A.1). The forced solution is sought in the form

$$u(x,t) = W(t) + \phi_1(t) \sin \frac{\alpha_1 x}{\varrho}$$
(A.13)

where ϕ_1 is an undetermined function of time. A virtual displacement

$$\delta u = a \sin \frac{\alpha_1 x}{\varrho} \tag{A.14}$$

is imagined, and one proceeds to calculate the virtual work

- (1) done by the inertia forces
 - (a) in the string (W_{1a})
 - (b) on the end mass (W_{1b})
- (2) done by the elastic forces (W_2)
- (3) of the exciting force (W_3)
- (4) done by the damping forces
 - (a) in the string (W_{4a})
 - (b) on the end mass (W_{4b}) .

These quantities are calculated below, using u and δu from Eqs. (A.13) and (A.14).

Calculation of W_{1a}

$$W_{1a} = -\int_{0}^{\varrho} \mu A \frac{\partial^{2} u}{\partial t^{2}} \delta u \, dx$$
$$= -\frac{\mu A a \varrho}{\alpha_{1}} (1 - \cos \alpha_{1}) \ddot{W} - \frac{\mu A a \varrho}{4\alpha_{1}} (2\alpha_{1} - \sin 2\alpha_{1}) \ddot{\phi}_{1} \qquad (A.15)$$

Calculation of W_{1b}

$$W_{1b} = -M \frac{\partial^2 u}{\partial t^2} \delta u \bigg|_{x = \varrho}$$

= $-Ma(\ddot{w} + \ddot{\phi}_1 \sin \alpha_1) \sin \alpha_1$ (A.16)

Calculation of W_2

The strain energy V is given by

$$V = \frac{1}{2} \int_{0}^{k} AE \left(\frac{\partial u}{\partial x}\right)^{2} dx = \frac{AE\alpha_{1}}{8k} (2\alpha_{1} + \sin 2\alpha_{1}) \phi_{1}^{2}$$

The change in V during the virtual displacement is

$$\delta V = \frac{\partial V}{\partial \phi_1} \delta \phi_1 = \frac{AE\alpha_1}{4\ell} \left(2\alpha_1 + \sin 2\alpha_1 \right) \phi_1 \delta \phi_1$$

The work done is the negative of this quantity, with $\delta \phi_1$ replaced by *a*:

$$W_{2} = -\frac{AE\alpha_{1}a}{4\varrho} (2\alpha_{1} + \sin 2\alpha_{1}) \phi_{1}$$
 (A.17)

Calculation of W_3

The force acting at x = 0 is

$$\sigma A = AE \left. \frac{\partial u}{\partial x} \right|_{x = \varrho} = \frac{AE\alpha_1}{\varrho} \phi_1$$

The virtual displacement at x = 0, however, vanishes (the displacement is prescribed at x = 0), so the virtual work vanishes:

$$W_3 = 0$$
 (A.18)

Calculation of W_{4a}

$$W_{4a} = -\int_{0}^{\Re} D \frac{\partial u}{\partial t} \,\delta u \,dx$$

= $-\frac{Da\Re}{\alpha_1} (1 - \cos \alpha_1) \,\dot{W} - \frac{Da\Re}{4\alpha_1} (2\alpha_1 - \sin 2\alpha_1) \,\dot{\phi}_1$ (A.19)

Calculation of W_{4b}

$$W_{4b} = -C \frac{\partial u}{\partial t} \delta u \bigg|_{x = \varrho} = -Ca(\dot{w} + \dot{\phi}_1 \sin \alpha_1) \sin \alpha_1$$
(A.20)

The total work done during the virtual displacement must vanish in the physical system. Therefore, the expressions given by Eqs. (A.15) to (A.20) are summed and equated to zero. The present study is concerned only with harmonic excitation:

$$W(t) = W_0 \sin \Omega t \tag{A.21}$$

Making this substitution and performing the indicated operations, one has

$$\frac{1}{\beta c_1^2} \ddot{\phi}_1 + C \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_1} \sin 2\alpha_1 \right) + \sin^2 \alpha_1 \right] \dot{\phi}_1 + \frac{M \omega_1^2}{\beta} \left(\frac{1}{2} + \frac{1}{4\alpha_1} \sin 2\alpha_1 \right) \phi_1 = \\ = \frac{M W_0 \Omega^2}{\beta} \left[\frac{1}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \beta \sin \alpha_1 \right] \sin \Omega t - C W_0 \Omega \left[\frac{\gamma}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \sin \alpha_1 \right] \cos \Omega t$$
(A.22)

where

$$c_{1} = \left[M \left(\frac{1}{2} - \frac{1}{4\alpha_{1}} \sin 2\alpha_{1} + \beta \sin^{2} \alpha_{1} \right) \right]^{-1/2}$$
(A.23a)

$$\omega_1 = \frac{\alpha_1 c_0}{\varrho} \tag{A.23b}$$

$$c_0^2 = \frac{E}{\mu} \tag{A.23c}$$

$$\gamma = \frac{DR}{C}$$
(A.23d)

To discuss the solution of this differential equation for ϕ_1 , it is useful to digress momentarily and consider the canonical form

$$m\ddot{\phi} + c\dot{\phi} + k\phi = F_1 \cos\Omega t + F_2 \sin\Omega t \tag{A.24}$$

The general solution to this equation is given by a particular integral (i.e., any solution of the equation) plus the complete solution of the homogeneous equation (i.e., with $F_1 = F_2 = 0$). The latter solution is a transient one, decaying exponentially. After this transient dies out, the particular integral prevails, so, for the present purposes, only the particular integral need be considered. It can be verified by direct substitution that a particular solution of Eq. (A.24) is

$$\phi = \frac{F_1 \cos\left(\Omega t - \alpha\right) + F_2 \sin\left(\Omega t - \alpha\right)}{\sqrt{(k - m\Omega^2)^2 + c^2 \Omega^2}}$$
(A.25)

where

$$\alpha = \operatorname{Tan}^{-1} \left(\frac{c\Omega}{k - m\Omega^2} \right) \tag{A.26}$$

To find the maximum (steady-state) response, ϕ_m , one equates $\dot{\phi} = 0$, whereupon

$$\Omega t = \alpha + \operatorname{Tan}^{-1} \left(\frac{F_2}{F_1} \right) \tag{A.27}$$

Substituting this value for Ωt into Eq. (A.25), there results

$$\phi_m = \frac{\sqrt{F_1^2 + F_2^2}}{\sqrt{(k - m\Omega^2)^2 + c^2 \Omega^2}}$$
(A.28)

Returning now to the original problem, the maximum value of ϕ_1 (and hence of u) can be determined by using Eq. (A.28) where, for the present problem.

$$F_{1} \equiv -\dot{C}W_{0}\Omega \left[\frac{\gamma}{\alpha_{1}}(1-\cos\alpha_{1})+\sin\alpha_{1}\right]$$

$$F_{2} \equiv \frac{MW_{0}\Omega^{2}}{\beta} \left[\frac{1}{\alpha_{1}}(1-\cos\alpha_{1})+\beta\sin\alpha_{1}\right]$$

$$m \equiv \frac{1}{\beta c^{2}}$$

$$c \equiv C \left[\frac{\gamma}{4}\left(2-\frac{1}{\alpha_{1}}\sin2\alpha_{1}\right)+\sin^{2}\alpha_{1}\right]$$

$$k \equiv \frac{M\omega_{1}^{2}}{\beta} \left(\frac{1}{2}+\frac{1}{4\alpha_{1}}\sin2\alpha_{1}\right)$$

With these substitutions and Eq. (A.25), Eq. (A.13) becomes

$$\frac{u(x, t) = W(t) + (A.29)}{\frac{\beta}{\beta} \left[\frac{1}{\alpha_1}(1 - \cos\alpha_1) + \beta \sin\alpha_1\right] \sin(\Omega t - \alpha) - CW_0 c_1^2 \Omega \left[\frac{\gamma}{\alpha_1}(1 - \cos\alpha_1) + \sin\alpha_1\right] \cos(\Omega t - \alpha)}{\left\{\left[\frac{Mc_1^2 \omega_1^2}{\beta} \left(\frac{1}{2} + \frac{1}{4\alpha_1} \sin 2\alpha_1\right) - \frac{\Omega^2}{\beta}\right]^2 + C^2 c_1^4 \Omega^2 \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_1} \sin 2\alpha_1\right) + \sin^2\alpha_1\right]^2\right\}^{1/2} \sin\frac{\alpha_1 x}{\varrho}$$

In this expression, α is defined by Eq. (A.26) as

$$\alpha = \operatorname{Tan}^{-1} \left(\frac{c\Omega}{k - m\Omega^2} \right) = \operatorname{Tan}^{-1} \left\{ \frac{Cc_1^2 \Omega \beta \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_1} \sin 2\alpha_1 \right) + \sin^2 \alpha_1 \right]}{M \omega_1^2 c_1^2 \left(\frac{1}{2} + \frac{1}{4\alpha_1} \sin 2\alpha_1 \right) - \Omega^2} \right\}$$
(A.30)

To find the maximum response, it is useful first to consider the *relative* displacement v(x,t) = u(x,t) - W(t). Using Eq. (A.28), one finds the maximum, $v_m(x)$, to be given by

$$v_m(x) = Nc_1 \sin \frac{\alpha_1 x}{\varrho} \tag{A.31}$$

where

$$N = F \left\{ \left[\frac{M\omega_1^2 c_1^2}{\beta} \left(\frac{1}{2} + \frac{1}{4\alpha_1} \sin 2\alpha_1 \right) - \frac{\Omega^2}{\beta} \right]^2 + C^2 \Omega^2 c_1^4 \left[\frac{\gamma}{4} \left(2 - \frac{1}{\alpha_1} \sin 2\alpha_1 \right) + \sin^2 \alpha_1 \right]^2 \right\}^{-1/2}$$
(A.32)
$$F^2 = W_0^2 c_1^2 \left\{ \frac{M^2 \Omega^4}{\beta^2} \left[\frac{1}{\alpha_1} (1 - \cos \alpha_1) + \beta \sin \alpha_1 \right]^2 + C^2 \Omega^2 \left[\frac{\gamma}{\alpha_1} (1 - \cos \alpha_1) + \sin \alpha_1 \right]^2 \right\}$$
(A.33)

The maximum dynamic stress is computed using Eq. (A.31):

$$\sigma_m(x) = E\epsilon(x) = E \frac{\partial u}{\partial x} = E \frac{\partial v}{\partial x}$$

$$\sigma_m(x) = \frac{ENc_1\alpha_1}{\varrho} \cos\frac{\alpha_1 x}{\varrho}$$
(A.34)

and the peak dynamic stress is seen to act at the upper end of the drill string.

The maximum *absolute* displacement, $u_m(\ell)$, is found by seeking stationary values of Eq. (A.29) with respect to time. This equation is first expressed in the more convenient form

$$u(\ell,t) = W_0 \sin \Omega t + \frac{N}{F} \left[T_1 \sin \left(\Omega t - \alpha \right) - T_2 \cos \left(\Omega t - \alpha \right) \right]$$
(A.35)

where

$$T_1 = \frac{MW_0 c_1^2 \Omega^2}{\beta} \left[\frac{1}{\alpha_1} \left(1 - \cos \alpha_1 \right) + \beta \sin \alpha_1 \right] \sin \alpha_1$$
(A.36a)

$$T_2 = CW_0 c_1^2 \Omega \left[\frac{\gamma}{\alpha_1} (1 - \cos \alpha_1) + \sin \alpha_1 \right] \sin \alpha_1$$
(A.36b)

On equating the first-time derivative to zero, one finds

$$\Omega t = -\operatorname{Tan}^{-1} \left[\frac{(FW_0/N) + T_1 \cos \alpha + T_2 \sin \alpha}{T_1 \sin \alpha + T_2 \cos \alpha} \right]$$
(A.37)

Substituting (A.37) into (A.35) leads to the expression for $u_m(\ell)$:

$$u_m(\ell) = \left\{ \left[W_0 + \frac{N}{F} \left(T_1 \cos \alpha - T_2 \sin \alpha \right) \right]^2 + \frac{N^2}{F^2} \left(T_1 \sin \alpha + T_2 \cos \alpha \right)^2 \right\}^{1/2}$$
(A.38)

This completes the derivation of all of the basic dynamic heave response equations stated as Eqs. (2.4) to (2.17) in Section 2.

APPENDIX D

Hydrogen Movement in Steel-Entry, Diffusion, and Elimination

HYDROGEN MOVEMENT IN STEEL-ENTRY, DIFFUSION, AND ELIMINATION

DEFENSE METALS INFORMATION CENTER Battelle Memorial Institute Columbus, Ohio 43201

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The Defense Metals Information Center was established at Battelle Memorial Institute at the request of the Office of the Director of Defense Research and Engineering to provide Government contractors and their suppliers technical assistance and information on titanium, beryllium, magnesium, aluminum, refractory metals, high-strength alloys for high-temperature service, corrosion- and oxidation-resistant coatings, and thermal-protection systems. Its functions, under the direction of the Office of the Director of Defense Research and Engineering, are as follows:

- 1. To collect, store, and disseminate technical information on the current status of research and development of the above materials.
- 2. To supplement established Service activities in providing technical advisory services to producers, melters, and fabricators of the above materials, and to designers and fabricators of military equipment containing these materials.
- To assist the Government agencies and their contractors in developing technical data required for preparation of specifications for the above materials.
- 4. On assignment, to conduct surveys, or laboratory research investigations, mainly of a short-range nature, as required, to ascertain causes of troubles encountered by fabricators, or to fill minor gaps in established research programs.

Contract No. AF 33(615)-1121 Project No. 8975

Roger & Runck

Roger J. Runck Director

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HYDROGEN MOVEMENT IN STEEL--ENTRY, DIFFUSION, AND ELIMINATION

by

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to

OFFICE OF THE DIRECTOR OF DEFENSE RESEARCH AND ENGINEERING

DEFENSE METALS INFORMATION CENTER Battelle Memorial Institute Columbus, Ohio 43201

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"The Problem of Hydrogen in Steel" DMIC Memorandum 180, October 1, 1965

"Hydrogen-Induced, Delayed, Brittle Failures of High-Strength Steels" DMIC Report 196, January 20, 1964

"The Effects of High-Pressure, High-Temperature Hydrogen on Steel" DMIC Report 202, March 26, 1964 By

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SUMMARY

Previous DMIC reports have described various problems associated with hydrogen in steel. The present report considers how hydrogen enters steel, moves through it, and is removed from it.

The solubility of hydrogen in iron and steel affects the entry of hydrogen into the material and its movement through the metal. The solubility also serves as an indication of the magnitude of the concentration gradients to be expected. Therefore, an understanding of the solubility of hydrogen is important in understanding other aspects of the behavior of hydrogen in steel.

Iron is a typical endothermic occluder of hydrogen. It is generally concluded that no hydrides are formed. The hydrogen atom, being the smallest atom, is the one most likely to form interstitial solid solutions. Apparently, it does so in iron, but X-ray data show no clear-cut evidence of solid solubility. The solubility of hydrogen in iron generally conforms to Sieverts' law for pressures up to at least 100 atmospheres. Therefore, it is concluded that hydrogen enters iron and steel only in the dissociated form, not as molecules. However, it is still uncertain whether this dissociated form is the atom or the proton, or both. Because the interstices in the α -iron lattice are smaller than those in v-iron, the solubility is lower in α -iron than it is in γ -iron. Because the interstices in α -iron are even smaller than the hydrogen atom, severe distortion of the α -iron lattice results from interstitial solution. The lattice expansion apparently is below the limit of experimental error for detection by X-ray methods. However, the bowing of a sheet specimen cathodically charged with hydrogen on only one surface and the elongation of a coil of iron wire upon being charged with hydrogen have demonstrated that ex pansion does accompany hydrogen solution.

At room temperature, the solubility of hydrogen in iron is very small (a few ppm), but the solubility increases with increasing temperature. A marked increase in solubility accompanies the transformation from body-centered cubic iron to the face-centered form on heating. Also, a large increase occurs on melting, the solubility in liquid iron being much greater than in the solid. In conformance with Sieverts' law, the solubility generally is proportional to the square root of the external partial pressure of molecular hydrogen. Alloying elements have some effect on solubility, but the crystal structure has a far greater effect.

Measurements of the solubility (S), permeability (P), and diffusivity (D) of hydrogen in iron are individually reliable and, under conditions of steady-state diffusion, they are consistent with each other at temperatures above about 750 F (that is, they satisfy the relationship $P = S \cdot D$), but not at lower temperatures. Experimental difficulties have precluded obtaining as complete an understanding of the solubility of hydrogen in iron and steel for the temperature range from about 750 F to room temperature as has been obtained for the higher temperatures. Nearly all investigators have obtained abnormally high solubilities of diffusible hydrogen at temperatures below about 750 F as compared with the extrapolation of data for α -iron at higher temperatures. There is strong evidence for the existence of traps in annealed material (as well as in cold-worked material) to account for the extra solubility. However, the dislocation density of annealed iron is far too low for the extra hydrogen to be trapped in dislocations. Some investigators suggest that other lattice defects, possibly vacancies, constitute the traps. Others suggest that the traps may be merely slightly widened planar lattice spacings. One group of investigators found that certain types of iron and steel treated at high pressure and high temperatures for prolonged periods will retain at ambient temperatures and pressures greater amounts of hydrogen than the usually indicated solubility. Furthermore, the hydrogen thus introduced can be extracted in two fractions, the "extra" hydrogen being much more difficult to remove. In this lowtemperature range, austenitic materials have higher solubilities for hydrogen than do the ferritic or martensitic materials, just as at higher temperatures.

A number of investigators have shown that cold work or subsequent annealing have little or no effect on hydrogen solubility in high-purity iron. However, in the case of SAE 1020 steel, for example, cold work may increase the solubility for hydrogen by a factor of as much as 100 or more for some conditions. Variations from this behavior

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have been reported, and there is no agreement even on the trends. Some of these discrepancies probably arose because of the use of different experimental conditions and because of variations in the methods used for the analysis for hydrogen, all of which are not equally effective in detecting both dissolved and trapped hydrogen. If the data indicate a general trend, it seems to be that cold work increases the solubility for hydrogen. Such increasing solubility is usually explained by hypothesizing the existence of "traps" in which that hydrogen in excess of the equilibrium lattice solubility is accommodated. Presumably, the concentration of such traps increases with increasing degree of plastic deformation. Apparently all such traps cannot be removed by annealing, but usually the large occlusive capacity of cold-worked material is markedly reduced by annealing.

Because the equilibrium solubility of hydrogen decreases with decreasing temperature, hydrogen tends to escape from surfaces, both external and internal. Atomic hydrogen depositing in internal voids will combine to form molecular hydrogen which cannot diffuse in steel, and thus high pressures of molecular hydrogen can build up in these internal voids, sometimes exceeding the yield strength of the steel so that blisters or other damage occurs.

The permeation of hydrogen through steel consists of interactions at both the entry and exit surfaces of the metal as well as diffusion through the bulk metal. Where diffusion is rate controlling and surface resistances to entry and exit are very small, permeation is equal to the product of the solubility and the diffusivity ($P = S \cdot D$). This relationship usually applies at temperatures above about 750 F.

Hydrogen can be introduced by corrosion processes, including acid pickling, corrosion in aqueous environments, and sulfide corrosion; electrochemical processes, such as electroplating, cathodic cleaning, electrolytic machining, and cathodic protection; exposure to high-pressure hydrogen or to hydrogen or water vapor at elevated temperatures; cyclotron proton irradiation; exposure to suitable flames; and glow discharge. Hydrogen introduced during steelmaking operations frequently is a factor in the delayed hydrogen cracking of steel. The report considers possible hydrogen-evolution reactions during corrosion or electrolytic processes; then the entry mechanism is discussed. Some of the important factors in hydrogen entry are the type of environment at the entry surface; the nature and concentration of the pickling bath, electrolyte or electroplating bath; pH, external hydrogen pressure, or current density; the presence of cathodic poisons or pickling inhibitors; the structure of the steel; and the condition of the steel surface, cold work having a significant effect on hydrogen entry. A number of substances, such as phosphorus, sulfur, hydrogen sulfide, arsenic, and selenium (called cathodic ii poisons) serve to delay the recombination of atomic hydrogen liberated at the steel surface into the molecular form, and, hence, provide increased opportunity for atomic hydrogen to enter the steel.

Hydrogen pickup in steelmaking is minimized by careful drying of raw materials, slagmaking materials, and refractories, and by such processes as vacuum melting or vacuum degassing. Heat-treating atmospheres should be low in hydrogen partial pressure and in water-vapor content to minimize hydrogen pickup. No acid pickling, cathodic cleaning, or electroplating operations can be tolerated if the hydrogen content of steel is to be kept as low as possible.

The second major part of permeation is diffusion through the bulk solid. Most of the current theories explaining hydrogen embrittlement and hydrogen cracking depend on the diffusion of hydrogen. The diffusion of hydrogen in iron appears to be compatible with the relation $P = S \cdot D$ at elevated temperatures, but anomalously low values of diffusivity are obtained at the lower temperatures, primarily below about 300 or 400 F. The results of various investigators in determining the diffusion coefficient are in fairly good agreement for temperatures above 750 F and generally so above about 400 F. However, experimental difficulties in measuring low rates of permeation have resulted in considerable discrepancy at temperatures near room temperature.

Experiments have shown that the diffusion of hydrogen in iron and steel occurs throughout the bulk material and not preferentially at grain boundaries as it does for many other materials. At higher temperatures, apparently hydrogen diffuses through the crystal lattice; at lower temperatures where anomalous solubility and diffusivity are encountered, it is possible that hydrogen diffuses through a nonconnecting "rift" network. As would be expected, the rate of diffusion increases rapidly with increasing temperature. This is the basis for the baking treatments designed to remove hydrogen from steel. Atomic hydrogen in steel will diffuse against very high back pressures of gaseous molecular hydrogen. Diffusion occurs at a barely perceptible rate at the temperature of dry ice (-109 F), and the rate is nil at the temperature of liquid nitrogen (-320 F).

Alloy additions of silicon and chromium to steel markedly decrease the diffusivity in α -iron. Increasing the carbon content appreciably lowers the permeation rate at ambient temperatures. Steel microstructure has a marked effect on the permeation of hydrogen through steel. Carbide particles in any form appear to hinder passage of hydrogen through steel, even after steady-state conditions have been attained. High permeation rates are obtained when the iron carbide is in the globular (spheroidized) form, and minimum rates are associated with martensite. Annealed or normalized structures have appreciably higher rates than do martensitic structures. Coarse, lamellar pearlite offers more resistance to hydrogen passage than does fine pearlite. Where permeation is determined by the rate of diffusion and not by surface processes on the entry or exit surfaces, the rate of permeation is inversely proportional to the metal thickness.

Controversial results have been obtained in studying the effect of cold work on the occlusion of hydrogen and also on permeation. Some investigators have reported that cold work increases the rate of permeation of hydrogen, others state that it reduces permeation, and still others have observed no effect or different effects depending upon the amount of cold work. Much more work is needed to clarify the effect of cold work on permeation and on diffusion.

The rate of hydrogen removal from iron and steel depends on, and varies greatly with, such factors as temperature, thickness of material, crystal structure (whether austenitic or ferritic), the amount of cold work, the environment at the exit surface, and the presence of coatings such as oxide films or electroplates.

The loss of hydrogen from austenitic steels is imperceptible at room temperature, but ferritic or martensitic steels continually lose hydrogen at ambient temperatures until the equilibrium level has been reached. The rate of loss from all steels increases with increasing temperature, at least within a few hundred degrees of room temperature. Hydrogen removal is roughly 250 to 500 times faster at 400 F than at room temperature. However because of increased solubility of hydrogen in α -iron with increasing temperature, baking temperatures should not be too high. The optimum temperature is about 400 to 500 F, but frequently it must be limited to 375 to 400 F to prevent damage to electroplated coatings. Variation in crosssectional area has a large effect on diffusion, and hence on the removal of hydrogen. For a cylindrical bar, the time to remove a given fraction of the hydrogen from a saturated specimen increases as the square of the diameter. In one instance, where 60 days was required to reduce the hydrogen to the equilibrium level in 1-inch-square bars at room temperature, about 6 years was required for 4inch bars.

Practically all alloy additions retard the evolution of hydrogen from steel, silicon and chromium being particularly effective. The method of introducing hydrogen into the steel has practically no effect on the rate of removal of hydrogen or the rate of recovery of lost ductility.

Frequently, surface reactions, rather than diffusion through the bulk of the metal, control the entry and removal of hydrogen. The nature of the environment at the exit surface has been found to be very important in the removal. For example, immersion in hot water is more effective than exposure to hot air at the same temperature; moist air is more effective than dry air. Oxide films, which often form on steels during treatment, hinder the removal of hydrogen. The presence of a more or less impermeable coating, such as dense cadmium, makes the evolution of hydrogen from steel more difficult, and frequently baking is rather ineffective in reducing the propensity toward delayed, brittle failure when such coatings are present. Ductility is recovered much more slowly in a baking treatment when coatings are present than when the steel surface is bare.

Residual or so-called "nondiffusible" hydrogen has been encountered many times in studies of hydrogen removal and in the analysis of steels for hydrogen. Frequently, the hydrogen can be extracted in two fractions, the one fraction requiring very high temperatures for its removal. This has led some investigators to suggest that hydrogen exists in steel in two forms, the one form being bound far more tenaciously than is indicated by the usual diffusion coefficient. Frequently, all of the hydrogen is not removed in warm-extraction analyses, but drastic vacuum treatment at temperatures in the austenitic region or even in the molten state is required.

Thus, removal is facilitated by elevated temperatures, clean surfaces free from oxide or other coatings, thin sections, low partial pressures of dissociated hydrogen in the external atmosphere, and by the presence of water or water vapor at the exit surface.

INTRODUCTION

The effects of hydrogen on the ductility of steel and the phenomenon of hydrogen-induced delayed, brittle failure of high-strength steel at low applied stresses have been described in a previous report (DMIC Report No. 196). It was shown that these phenomena depend not only on the presence of hydrogen, but also on the ability of hydrogen to diffuse, or move, in the steel while it is under load. This movement occurs either in response to a concentration gradient or a stress gradient. Since both the presence and the movement of hydrogen in steel are fundamental to these serious problems, this report has been prepared to aid in understanding the movement of hydrogen in steel. It considers the ways in which hydrogen enters steel, how it moves through steel, and methods whereby it may be removed from steel. The various factors that affect each of these phenomena are considered.

The person who is producing high-strength steel parts is interested in keeping hydrogen out of steel to as great an extent as possible, and in being able to remove most of that which enters unavoidably. Sometimes the research man desires to introduce hydrogen into steel intentionally in order to study the influence of hydrogen on the behavior of steel. Information is given to assist in both of these approaches to the hydrogen problem.

First, the report deals with the solubility of hydrogen in iron and steel. Following this, the permeation of hydrogen through iron and steel is broken down into its component parts of entry, diffusion, and exit. A concluding section deals with the removal of hydrogen from production parts.

SOLUBILITY

Introduction

The effect of hydrogen on the mechanical properties of any given metal depends upon the hydrogen-metal system, and, especially, on such factors as the solubility of hydrogen in the metal and the formation of a hydride phase. It is generally considered that metals and hydrogen react in one of the four following ways: (1) by forming salt-like, ionic hydrides, (2) by forming covalent hydrides that are gaseous at room temperature, (3) by forming alloys that exhibit negative heats of absorption (endothermic occlusion), or (4) by forming alloys that exhibit positive heats of absorption (exothermic occlusion), sometimes called "pseudo hydrides".

Iron, being a typical endothermic occluder, takes up hydrogen only in small quantities, even at elevated temperatures where such occluders exhibit their greatest solubilities. This behavior is in accordance with the positive temperature coefficient which, of necessity, corresponds to the negative heat of absorption. The alloy of iron and hydrogen, that is, the product of occlusion, consists of only one solid phase. This was long considered to be a true terminal solid solution of hydrogen in iron. Proponents of the viewpoint that endothermic occluders form true solid solutions with hydrogen cite the following as evidence: (1) the hydrogen solubility varies with the square root of the pressure (Sieverts' law), and (2) the solubility increases as the temperature increases. (1) Other investigators question whether solid solutions occur at all in endothermic metal-hydrogen systems, such as the iron-hydrogen system.⁽²⁾ As will be shown later, the solubility of hydrogen in iron does not follow Sieverts' law perfectly, X-ray evidence, including some obtained recently, (3) fails to indicate solid solubility.

The exothermic occluders, which absorb hydrogen in much larger quantities than do the endothermic occluders, form alloys of greater complexity which exhibit unmistakable solid solutions and one or more secondary phases.

Although the products of occlusion of the exothermic occluders and endothermic occluders of the transition metals resemble the hydrides of the alkali and alkaline-earth metals in showing variable composition, the analogy appears to extend no further; the hydrogen alloys form either with expansion or without any volume change at all and are wholly metallic in character, while formation of the hydrides of the alkali and alkaline-earth metals is accompanied by marked contraction and the products are salt-like in nature. The changes in gross dimensions, brought about by absorption of hydrogen, are, in the cases of endothermically occluding metals (like iron), generally too small to be observed with certainty. The exothermically occluding metals, on the other hand, are expanded

markedly by occlusion, as much as 15 percent or more.

Knowledge of the solubility of hydrogen in iron and steel is important in understanding the hydrogen problem, because the solubility affects the entry of hydrogen into the metal and also its movement through the metal. It serves to indicate the magnitude of the concentration gradients to be expected.

It will be shown later, in the section on permeability, that the measurements of solubility, permeability, and diffusivity of hydrogen in iron are individually reliable and, under conditions of steady-state diffusion, they are consistent with each other for temperatures above 750 F (400 C). However, at temperatures below about 750 F, the experimentally measured solubility deviates from the extrapolation of the high-temperature behavior in the direction of greater solubility. Because of these apparently anomalous results, the solubilities at high temperatures and low temperatures will be discussed separately.

The gas content in a solid or liquid may be expressed in a number of ways. The commonest are as follows:

- Atomic concentration = atoms of H per atom of metal
- Relative volume (RV) = volumes of H per volume of metal
- Weight percent = g of H per 100 g of metal
- Gas volume per metal weight = commonly, cc of H per 100 g of metal, or, less often, cc of H per g of metal
- Parts per million by weight (ppm) = 0.0001 x weight percent.

Conversion factors for interconversion of these units for iron are listed in Table 1.

Preferred Lattice Sites for Hydrogen

Interstitial solid solutions are formed when solute atoms of small radii are accommodated in the interstices, or "holes", of the lattice of a solvent. Because the size of the lattice interstices is restricted, only the smaller atoms hydrogen, boron, carbon, nitrogen, and oxygen are likely to form this type of solid solution. Of these, the hydrogen atom with an atomic radius of 0.46 Å is the smallest, and, hence, the most readily accommodated interstitially. When a solvent accepts one of these elements interstitially, solution is usually accompanied by expansion of the unit cell.

Diagrams of both of the iron lattices are shown in Figure 1. The largest holes in the

TABLE 1. FACTORS FOR CONVERSION OF DIFFERENT UNITS OF HYDROGEN CONTENT IN IRON

(Metal density assumed 7.88 g/cc; atomic weight of iron 55.84.)

Factor												
Atomic Concentration	RV	Weight Percent	Cc H2 100 g Fe	Ppm								
1	0,000632	0,554	0,0000498	0,0000554								
1,580	1	877	0.0788	0.0877								
1,805	0.00114	1	0.0000899	0.0001								
20,080	12.69	11,100	1	1,11								
18,050	11.4	10,000	0.899	1								
	Concentration 1 1,580 1.805 20,080	Concentration RV 1 0.000632 1,580 1 1.805 0.00114 20,080 12.69	Atomic Concentration Weight RV Weight Percent 1 0.000632 0.554 1,580 1 877 1.805 0.00114 1 20,080 12.69 11,100	Atomic Concentration Weight RV Cc H ₂ Percent 1 0.000632 0.554 0.0000498 1,580 1 877 0.0788 1.805 0.00114 1 0.0000899 20,080 12.69 11,100 1								

Example of calculation:

Calculate the atomic concentration (H/Fe) when RV = 94 94 x 0.000632 = 0.0594, i.e., about 0.06 atom H per 1 atom Fe.

close-packed face-centered cubic lattice are at the midpoints of the edges of the unit cell $(00\frac{1}{2},$ etc.) and at $\frac{1}{2}\frac{1}{2}$. These are known as octahedral sites and are crystallographically equivalent positions; one example of the former type of site and the only latter site are illustrated in Figure la. Each of these interstices is surrounded symmetrically by six iron atoms (radius 1.26 Å) at equivalent distances, giving a coordination number of 6. Smaller interstices exist at positions of the type $\frac{1}{4}\frac{1}{4}\frac{1}{4}$; these are surrounded by a tetrahedral arrangement of four solvent atoms (coordination number = 4), and hence, are known as tetrahedral sites. One of these sites is shown in the upper right portion of Figure 1a. In y-iron, the larger (octahedral) interstices have room for a spherical atom of 0.52 Å radius, and the smaller (tetrahedral) interstices for one of only 0.28 Å without expansion of the lattice. The kind of site occupied by a given interstitial atom is that which will permit it to attain the highest coordination number compatible with its size. Larger interstitial atoms tend to occupy octahedral sites and smaller ones the tetrahedral sites, the maximum radius ratio being about 0.59 for CN = 6 and 0.41 for CN = 4. It is considered that the preferential location of the hydrogen atom is the smaller (tetrahedral) site of the face-centered cubic y-iron lattice⁽⁴⁾ the H/Fe radius ratio being only 0.37.

Body-centered cubic metals have their largest holes at $\frac{1}{2}\frac{1}{4}0$ and equivalent positions. A spherical atom here would touch four spherical atoms of the solvent. One of these sites is shown in Figure 1b. In α -iron, there would be room in these tetrahedral sites for an interstitial atom of 0.36 A radius, which is smaller than even the hydrogen atom. Smaller holes are found at the midpoints of the edges ($00\frac{1}{2}$, etc.) and at the equivalent positions at the centers of the faces $(\frac{1}{2})_{2}^{1}$, etc.). One of each of these small, octahedraltype sites is indicated in Figure 1b. These octahedral sites are surrounded by two close atoms and four slightly more distant ones and, thus, are asymmetrical (see points denoted "A" in the figure). In α -iron, there would be room here for a sphere of only 0.19 Å radius if the iron atoms are con-/167



 a. Interstitial Positions in Face-Centered Cubic y-Iron



b. Interstitial Positions in Body-Centered Cubic α-Iron

FIGURE 1. SCHEMATIC REPRESENTATION OF α - AND γ -IRON LATTICES

(Solid circles denote centers of iron atoms.)

sidered as spheres in contact with each other. The larger (tetrahedral) sites in α -iron are regarded to be more stable for hydrogen than are the asymmetrical positions. (5,6)

Because the interstices in α -iron are smaller than those in γ -iron, it appears that interstitial solid solution in α -iron should be more difficult than in γ -iron, as is the case. Because the interstices in α -iron are considerably smaller than even the hydrogen atom, one concludes that severe distortion of the α -iron lattice should result from interstitial solution.

Plusquellec and co-workers(7) have suggested that hydrogen is preferentially associated with 3

lattice defects present in the [112] crystal planes. From X-ray studies, Bastien⁽⁸⁾ concluded that the $\{112\}$ crystal planes in α -iron are tilted slightly in the presence of hydrogen, but the parallelism of the planes is restored as the hydrogen diffuses out of the material. This conclusion was based on the observed broadening of the [112] spots on Laue patterns obtained from steel charged with hydrogen. Heating the previously charged steel at temperatures high enough to drive the hydrogen out but too low to cause any recovery of the lattice caused the broadening to disappear. He attributed this effect to a local lattice distortion associated with hydrogen occupying and expanding the tetrahedral lattice interstices of body-centered cubic a-iron. Recently, Tetelman, Wagner, and Robertson⁽³⁾ showed by X-ray studies that cathodic charging of iron with hydrogen produces both distortion and anisotropic particle-size broadening. They concluded that the broadening of the X-ray reflections, that was found after charging, resulted from plastic deformation and was not due to preferential location of hydrogen in the lattice as Plusquellec and co-workers assumed.

Since, as will be shown later, the solubility of hydrogen in iron conforms to Sieverts' law at temperatures above 750 F (400 C), it has been concluded that hydrogen enters iron in a dissociated form, rather than as molecules. Furthermore, the relatively simple diffusivity behavior of hydrogen in iron at temperatures above 750 F (400 C) has been taken as indicating that hydrogen diffuses interstitially at these temperatures. This approach is verified by a consideration of interstitial diffusion theory in which Wert and Zener⁽⁹⁾ obtained theoretical values of 1.6×10^{-3} cm²/sec for each of two interstitial models in which it was assumed that the hydrogen occupied either the octahedral or the tetrahedral sites of α -iron. This value compares favorably with the experimental value of $1.4 \times 10^{-3} \text{ cm}^2/\text{sec}$ for the diffusion coefficient D_o obtained recently by Johnson and Hill⁽¹⁰⁾ for temperatures above 390 F (200 C). This leaves little doubt as to the interstitial nature of the diffusion process, for, according to Zener, interstitial processes are the only diffusion processes in which the theoretical D_0 is of the order of 10^{-3} cm²/sec. However, this method could not determine which site was preferred.

Although it has been demonstrated that hydrogen enters and diffuses through iron in a dissociated form, it is still uncertain whether this form is an atom or a proton. The extremely rapid diffusion rate of hydrogen in iron has led some to speculate that the dissociated particles are protons which have an extremely small radius (10^{-5} Å) with respect to the interatomic distance in iron. Bastien⁽⁸⁾ has presented evidence from electrolysis experiments which indicate that the hydrogen is present as ionized particles with a positive charge. He has speculated that the hydrogen probably does not exist as simple protons, but rather as protons that are partially shielded or screened by electrons. The effective size of such an ion could be near that of the hydrogen atom, big enough to conceivably lead to some lattice distortion and expansion. However, the state of ionization is not yet known. According to Bastien, (8) Darmois(11) has obtained additional evidence that the diffusion takes place at least partly in the form of protons.

Cottrell has shown that, for larger interstitial atoms, such as carbon and nitrogen, a significant decrease in lattice strain energy accompanies the transfer of an atom from an isolated interstitial site into one in the vicinity of a dislocation. This decrease in energy provides the driving force for the formation of atmospheres of such atoms around dislocations. Heller⁽¹²⁾ studied the hydrogen yield point and concluded that the corresponding strain interaction of hydrogen with dislocations is too small to lead to condensation around dislocations at room temperature.

Hill and Johnson(13) believe that the extremely rapid decrease in the diffusivity of hydrogen in annealed iron at temperatures below 390 F (200 C) is related to the existence of traps in which the hydrogen possesses lower energy than when it is in interstitial sites. Accordingly, diffusion at lower temperatures is presumed to be controlled by the rate of escape from such traps, thus accounting for the higher activation energy observed at these temperatures.

Hill and Johnson⁽¹³⁾ have presented the following views on the physical nature of the traps for hydrogen. The traps in unalloyed iron that has not been cold worked seem qualitatively similar in certain respects to those in cold-worked iron-carbon alloys, which these investigators studied previously and which they suggested were cracks formed by piled-up dislocations in which the hydrogen is chemisorbed. (14) The present traps react with hydrogen endothermically, however, and so they do not appear to be chemisorption sites, in view of the fact that hydrogen chemisorption is exothermic in all known cases. Nevertheless, there might be some connection between the traps in α -iron and those that result from cold work in steel. During plastic straining of steel, the present traps might migrate through the ferrite lattice and coalesce at interfaces with Fe₃C particles to form the larger and more energetic traps characteristic of deformed steel. They would thus serve as building blocks for the stronger traps in cold-worked steel.

This behavior of the traps suggests that they are dislocations. However, previous estimates of the strain interaction of hydrogen with dislocations are of the order of only 0.1 ev or 2,000 cal/g-atom. Such interactions would be detectable only at quite low temperatures, whereas the present trapping is detectable at 555 F (290 C) and the interaction energy is 4,800 cal/g-atom. Hill and Johnson suggest that the hydrogen may be at least partially ionized and, therefore, has an electrostatic rather than a strain interaction with the dislocations. Also, a dislocation

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density several orders of magnitude higher than the usually accepted dislocation density in undeformed iron would be required if the traps are actually dislocations. Likewise, in the temperature range of interest, the concentration of vacancies is several orders of magnitude smaller than the number of trapped hydrogen atoms.

It is possible that the traps are another kind of lattice defect. This has been suggested in effect in many previous theories of hydrogen in metals. Smith⁽²⁾has hypothesized that the hydrogen resides entirely in a network of interconnecting "rifts" created by the hydrogen itself. Hill and Johnson's view differs from that of Smith in holding that the hydrogen diffuses through the lattice, rather than through a rift network, and is situated wholly in the lattice interstices at higher temperatures. The possibility of nonconnecting "rifts" is admitted only at lower temperatures where the anomalous solubility and diffusivity (to be discussed later) are found. Other ideas of extra-lattice accommodation of hydrogen in steel include the planarpressure occlusion theory of Zapffe(15,16,17) and the theory that hydrogen is accommodated at subgrain boundaries suggested by Andrew et al. (18,19)

Hill and Johnson⁽¹³⁾ pointed out that all these views of hydrogen in metals have many points in common with their concept of traps. They suggested that a generally consistent conclusion might be that the traps are slightly widened planar lattice spacings that form only as the hydrogen-saturated iron lattice is cooled below a certain temperature. Such a mechanism does not require the presence of dislocations or vacancies, and there is virtually no limit to the number of traps that can be formed if the dissolved hydrogen concentration or the external pressure of molecular hydrogen is high enough.

Lattice Expansion

The exothermic hydrogen occluders exhibit considerable expansion of the crystal lattice as the result of hydrogen occlusion. From the data on iron available to $Smith^{(2)}$ in 1948, he concluded that expansion occurring as a result of hydrogen absorption is limited to exothermic occluders. However, it was shown in 1960 that this is not the case, because the face-centered cubic lattice of the endothermic occluder nickel is expanded by 6 percent as the result of cathodic charging with hydrogen.

Wever and Pfarr⁽²⁰⁾ reported an increase in the lattice spacing of mild steel of about 0.017 percent as the result of prolonged charging with cathodically evolved hydrogen. The average hydrogen content after charging was $30 \text{ cm}^3/100 \text{ g}$ of steel, itself a very high value, but the investigators believed that the hydrogen concentration in the surface layer was probably much higher. On standing at room temperature, the hydrogen gradually escaped and the lattice spacing was observed to decrease, approaching the initial value after 15 days. Baukloh and Stromburg⁽²¹⁾ observed a 0.03 percent increase in the lattice spacing as the result of cathodic charging for 24 hours under severe conditions. However, in other work^(22,23,3) no detectable influence of cathodic charging with hydrogen on the lattice spacing of iron was found. According to Smialowski⁽²⁴⁾ Lihl⁽²⁵⁾ observed changes in the lattice spacing similar to those reported by Wever and Pfarr. However, he concluded that high stresses arising in the surface layer of pickled or electrolytically charged specimens made a correction of the values of Wever and Pfarr mandatory. After making such a correction, he found that their values were below the limit of experimental error.

Some of the investigators who used X-ray methods for studying the effects of hydrogen occlusion in iron or steel tried to evaluate the magnitude of the stresses introduced in the surface layers of the metal by charging with hydrogen either electrolytically or by pickling. For example, Lihl(25) reported surface compressive stresses as high as 27,000 psi. However, he suggested that these stresses are a secondary effect of corrosion and cannot be directly attributed to the presence of occluded hydrogen in the metal. Glikman and Snezhkova⁽²⁶⁾ cathodically charged one side of a 13 percent chromium steel strip with hydrogen. The charged side became convex, and its deflection was measured. The highest stresses calculated were 10,000 psi. Inasmuch as the deflection of charged specimens disappeared after they were baked at about 400 to 575 F, these investigators concluded that the stresses were created by dissolved hydrogen, not by corrosion as Lihl suggested. Szklarska-Smialowska and Smialowski(27) also observed the bending of strips cathodically charged on one side only. They found that iron actually deformed plastically as the result of cathodic charging with hydrogen. Smialowski and Szklarska-Smialowska, (28) studying the elongation of iron wires as the result of cathodic charging with hydrogen, observed linear expansion when catalytic poisons were present in the electrolyte; the effect was a small fraction as great when no poison was added to the acid electrolyte. This work will be discussed further when the effect of poisons on hydrogen entry is considered. In studying the deformation of Cshaped specimens cut from cold-rolled steel tubing and charged only on the ID, the cold-worked steel did not become deformed under the influence of cathodic charging. However, with prior annealing. the width of the gap gradually expanded as cathodic charging progressed. The deformation was almost completely irreversible, the width of the gap decreasing very little on standing at room temperature. Metallographic examination of charged rings showed that blisters and cracks were present. After a certain period of electrolysis, both the wires and the rings reached a limiting degree of deformation, no further deformation being observed with continued electrolysis.
Foryst and Smialowski^(29,24) studied the limiting expansion and relative embrittlement of two Armco iron rods cathodically charged with hydrogen. One of the irons was unkilled and contained many sulfide and oxide inclusions; the other iron was Si-Al killed and had a much lower inclusion content. Some of the data for these two irons are included in the following tabulations:

	Iron			
	Unkilled	Killed		
Composition, weight				
percent				
C	0.02	0.03		
Mn	0.05	0.06		
Si	0.03	0.15		
P	0.007	0.011		
S	0.031	0.004		
Al	0.00	0.07		
Oxides	0.061	0.039		
Limiting Expansion,				
percent	0.021	0.004		
Relative Embrittlement,				
percent	58	60		

The unkilled iron with the high inclusion content showed about five times greater expansion when cathodically charged with hydrogen than did the killed iron. However, these irons did not differ in their susceptibility to embrittlement by cathodically evolved hydrogen. Thus, embrittlement and deformation or expansion do not seem to be quantitatively related to each other. The explanation advanced for this difference in behavior was that embrittlement depends on the degree of impregnation of the metal phase with atomic or ionic hydrogen, whereas expansion is due to the production of voids filled with gaseous hydrogen under high pressure. It was concluded that the presence of internal voids and defects is essential for the occurrence of elastic and plastic deformations in iron cathodes as the result of being charged with hydrogen. Embrittlement is a property of the metal phase, while expansion depends on the quantity and distribution of voids, nonmetallic inclusions, and other defects in the metallic structure. Since Foryst(30) later found that oxide inclusions did not influence the expansion of iron, it seems likely that the expansion attributed to inclusions is associated with sulfides.

The present authors have detected bowing of thin sheet specimens of a number of irons and mild steels cathodically charged with hydrogen on one surface. The bowing, measured with a dial indicator, was convex on the charged surface.

Although direct measurements of lattice expansion and stresses resulting from charging iron and steel with hydrogen have given controversial results, the bowing of sheet specimens charged on one side and the elongation of a spiral are evidence of expansion.

Measurements of Solubility at Elevated Temperatures

Effect of Temperature and Pressure

At room temperature, the solubility of hydrogen in iron and other endothermic occluders is very small. However, the solubility increases with increasing temperature, and sometimes becomes considerable at temperatures of 1200 F or above.

Several investigators have measured the solubility of hydrogen in annealed or hot-worked iron and steel. (31, 32, 33, 34, 35, 36, 37, 38, 13) Usually, the method employed was to measure the change in volume of gaseous hydrogen in contact with the sample at a given temperature and pressure.

At room temperature, the solubility of hydrogen in body-centered cubic α -iron is very small, but it increases slowly with increasing temperature. A marked increase in solubility accompanies the transformation from bodycentered cubic α -iron to face-centered cubic γ -iron at about 1670 F. The solubility in γ -iron also increases slowly with increasing temperature until the γ - \circ transformation temperature is reached, where the solubility drops markedly. In the body-centered cubic ô-form, the solubility again gradually increases with increasing temperature up to the melting point, where a large increase occurs. Thus, for a given temperature, the solubility of hydrogen in face-centered cubic iron is appreciably greater than that in body-centered cubic iron. This is a consequence of the larger interstices in the face-centered cubic lattice of y-iron as compared with those in the body-centered cubic α - and δ -iron lattices. Also, the solubility in liquid iron is much greater than that in solid iron. These relationships are shown by several solubility isobars in Figure 2. For a given isobar, the solubility of δ -iron may be considered as an extension of the solubility of α -iron, since both forms have the same crystal lattice.

In both the liquid and the solid states for iron and steel, the solubility of hydrogen generally follows Sieverts' law for diatomic gases and varies as the square root of the gas pressure:

$$S = k(p_{H_2})^{\frac{1}{2}}$$
 (1)

Actually, the solubility of hydrogen in steel is proportional to the partial pressure of <u>atomic</u> hydrogen, the only form of hydrogen that is soluble in steel. The partial pressure of atomic hydrogen formed by dissociation of molecular hydrogen is proportional to the square root of the pressure of molecular hydrogen.



FIGURE 2. Fe-H IRON-HYDROGEN CONSTITU-TIONAL DIAGRAM AFTER ZAPFFE(39)

Sieverts and co-workers showed the relationship expressed in Equation 1 to be generally applicable to endothermic hydrogen occluders, including iron, at all except high pressures and at all temperatures investigated (they were working at elevated temperatures). Armbruster⁽³⁷⁾ determined the solubility of hydrogen in iron, nickel, and several steels at low pressures that ranged from 0.001 to 1.5 mm Hg and temperatures of 750, 930, and 1110 F (400, 500, and 600 C), and verified the applicability of the pressure-solubility relationship given by Equation 1.

However, deviations from Sieverts' law have been observed, chiefly at lower temperatures than were used by Sieverts and by Armbruster. Hill and Johnson⁽¹³⁾ found positive deviations for the pressure exponent, n, from the Sieverts' law value of 0.5, as follows:

Tempe	rature	
С	F	n
585	1085	0.57
538	1000	0.54
490	914	0.52
390	734	0.63
290	554	0.55
210	410	0.56
145	293	0.75

Ham and Rast(40) reported that different specimens exhibited different values of n as high as 0.555 at 930 F (500 C); n increased with decreasing temperature to as high as 1.08 at 390 F (200 C). Chang and Bennett⁽⁴¹⁾ reported an increase from 0.5 to as much as 1.0 with decreasing temperature. Thus, in all these investigations, the solubility would appear to be anomalously high at low temperatures.

The solubility, S, as a function of temperature, T, was measured by Sieverts(32) for iron; Fowler and Smithells(42) showed that the results were described by an equation of the following type:

$$S = \alpha p^{1/2} e^{-\beta/T}, \qquad (2)$$

where α and β are constants. Armbruster(37) found that an equation of the same form accurately expressed the temperature relation of the solubility of hydrogen in pure iron in her experiments at temperatures in the range of 750 to 1110 F (400 to 600 C). What appears to be the best equation for the solubility of hydrogen in α -iron at temperatures above γ 50 F (400 C) is that given by Geller and Sun:(38)

$$S_{(ppm)} = 42.7 p^{\frac{1}{2}} e^{-3280/T}$$
. (3)

This equation is based on the average of several reliable measurements made independently by various investigators. (33, 34, 35) The results for a pressure of 1 atmosphere are shown graphically in Figure 3.



FIGURE 3. SOLUBILITY OF HYDROGEN IN PURE α - AND γ -IRON AS A FUNCTION OF TEMPERATURE(38)

 $(p_{H_2} = 1 \text{ atmosphere} = 760 \text{ torr.})$

Dyakonov and Samarin⁽⁴³⁾ computed equilibrium constants and energy changes for hydrogensolubility data obtained for iron by various investigators. When they plotted the logarithm of the equilibrium constant, K, for the reaction

$$\frac{1}{2}H_2$$
 gas $=$ [H] in Fe (4)

against 1/T (the term [H] being the equilibrium concentration of atomic hydrogen dissolved in iron under constant gas pressure), a straight-line relationship was found for each form of iron, including liquid iron. From the four straight lines obtained, they calculated the standard free-energy changes and the heats of hydrogen absorption for the various forms of iron - α , γ , δ , and liquid. The results showed that the values of the thermo-dynamic properties do not differ greatly for the different forms of solid iron.

Effect of Composition

Many investigators have studied the effects of different alloying elements on the solubility of hydrogen in iron and steel. Much of this work is for the molten state, and, hence, is of relatively little interest to those concerned with delayed, brittle failure of steel. This area has been reviewed by Smialowski(24) in Chapter 2. Only a few data available for the solid state will be presented to illustrate that there is an effect of composition on hydrogen solubility.

Armbruster(37) determined the solubility of hydrogen in iron, nickel, and 13 steels of vastly different composition, including both austenitic and ferritic steels. The samples are described in Table 2. Her results are shown in Table 3 and in Figure 4. S is the solubility in micromoles of hydrogen per 100 g of metal, and α is the slope of the plot of micromoles of hydrogen per millimeter of pressure versus the reciprocal of the square root of the pressure. From the solubility of hydrogen at 1.0 mm pressure in the steels at the various temperatures investigated, which is represented by the values of α in Table 3, it is apparent that the steels fall into three groups, namely, (I) ferritic, low in alloying elements, (II) an intermediate group, high in alloying elements, and (III) austenitic, the solubility of hydrogen in this group being several times as great as in the ferritic group. The magnitude of the differences among these groups is shown by the curves in Figure 4.

Luckemeyer-Hasse and Schenck⁽³⁵⁾ determined the solubility of hydrogen in iron, nickel, chromium, and a number of iron-nickel, ironchromium, and iron-nickel-chromium alloys. Their results are shown in Figures 5, 6, and 7.

Smialowski⁽²⁴⁾ has summarized many of the data regarding the effects of other alloying elements on hydrogen solubility in solid steel (see Chapter 2).

Effect of Crystal Structure

Although alloying elements have some effect on the solubility of hydrogen in iron, this effect is overshadowed by the influence of crystal structure on hydrogen solubility. It is clearly shown by Armbruster⁽³⁷⁾ that body-centered cubic iron and ferritic steels (Numbers 2 through 10) dissolve less hydrogen than do the face-centered cubic austenitic steels (Numbers 11 through 14) and nickel (see Figure 4 and Table 3). The results of Luckemeyer-Hasse and Schenck(35) that were presented above also support the same conclusion. These latter investigators apparently were the first to associate this effect of structure on the solubility of hydrogen with the size of the interstices, recognizing that there are larger empty spaces in the center of a model of a face-centered

TABLE 2. DESCRIPTION OF SAMPLES USED BY ARMBRUSTER(37)

					Composi	tion, per	rcent				Weight,	Thick- ness,	Area
Number	Material	С	Mn	S	Р	Si	Cr	Ni	0	ther	g	mm	Volume, cm ⁻¹
1	Carbonyl iron	0.011									516.6	0.34	65
2	Mild steel	0.39	0.73	0.032	0.022	0.19	0.048	0.012			439.6	2.45	10
3	Mild steel	0.39	0.70	0.049	0.016	0,128					382.1	0.076	255
4	Mild steel	0.12	0.47	0.041	0.016	0.003					366.7	0.038	530
5	Mild steel	0.10	0.39	0.032	0.015	0.007					499.5	0.27	80
6	1.5% Mn steel	0.20	1.46	0.022	0.027	0.205					510.7	0.26	82
7	3% Si Steel	0.05	0.23	0.013	0.010	3.17			A1	0.27	624.9	0.28	74
									A1,0	, 0.03			
8	4% Ni Steel	0.32	0.64	0.020	0.023	0,245		3.64	-	2	504.7	0.26	78
9	13Cr, 0.3C, ferritic	0.32	0.47			0.28	12.78				508.3	0.74	28
10	16% Cr steel	0.09	0.46	0.014	0.012	0.395	15.60	0.33			552.3	0.25	80
11	28% Ni steel	0.025	0.18			0.06		27.88			475.5	0.15	137
12	13% Mn steel	1.40	12.98	0.019	0.048	0.32					523.8	0.96	22
13	13Cr, 0, 3C, austenitic	0.32	0.47			0.28	12.78				508.3	0.74	28
14	18Cr-8Ni steel	0.07	0.37	0.005	0.006	0.47	18.30	9.92			362.2	0.097	209
15	Nickel	0.13	0.13			0.04		99.52	Cu Fe	0.06	377.7	0.14	152

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	micromol	les/100 g at	1.0 mm			
Sample Number	752 F (400 C)	932 F (500 C)	1112 F (600 C)	Total Number of Measurements	Average Deviation in Solubility	Heat of Solution (△H) cal/g-mole
1	0.60	1,16	1.93	108	±0.03	13,300
2	0.45	0.97	1.58	39	±0.02	14,200
3	0.79	1.29	2.18	74	±0.03	13,600
4	0.69	1.35	2,15	36	±0.04	13,100
5	0.45	1.00	1.79	36	±0.03	16,200
6	0.58	1.21	2.07	40	±0.04	14,800
7	0.46	1.52	2.42	43	±0.05	18,600
8	0.50	1.09	2.02	76	±0.02	16,200
9	0.33	0.73	1.46	38	±0.04	16,400
10	0.40	0.84	1.56	47	±0.03	15,900
11	1.51	2.44	3.54	48	±0.03	9,700
l2a (40%)	1.63	2.54	3.25	40	±0.04	8,000
12Ъ (85%)	2.45			16	±0.03	
13	4.37	5.20	6.18	29	±0.04	4,200
14	5.24	6.37	7.55	40	±0.03	4,300
15	5.93	7.94	9.83	45	±0.03	5,900

TABLE 3. VALUES OF α , THE SOLUBILITY OF HYDROGEN AT 1.0 MM PRESSURE, FOR CERTAIN STEELS AT 750 TO 1110 F (400 TO 600 C)⁽³⁷⁾



FIGURE 4. SOLUBILITY OF HYDROGEN AT 1 ATMOSPHERE IN IRON, NICKEL, AND A TYPICAL STEEL FROM EACH OF THREE GROUPS, AT 572 TO 1652 F (300 TO 900 C)⁽³⁷⁾

(S = solubility in micromoles of H per 100 g of metal.)



FIGURE 5. INFLUENCE OF TEMPERATURE ON THE SOLUBILITY OF HYDROGEN IN IRON, NICKEL, CHROMIUM, AND SEVERAL IRON-NICKEL ALLOYS(35)



FIGURE 7. DISSOLVING CAPACITY OF VARIOUS Fe-Ni-Cr ALLOYS FOR HYDROGEN WITH INCREASING TEMPERATURE(35)

Alloy	<u>C, %</u>	Ni, %	Cr, %
1	0.01	3.1	2.57
Z	0.01	8,12	4.43
3	0.01	11.4	3.02
4	0.01	26.4	2.79
5	0.01	5.1	4.65
6	0.01	10.0	4.63
7	0.01	5.16	9.53
8	0.01	20.3	9.66
9	0.01	15.1	14.6
10	0,01	10.0	29.5
11	0.02	2.04	44.1
12	0.02	5.1	44.2
13(a)	0.05	9.02	17.5

cube formed by spheres than in the body-centered cube. Sieverts and Hagen(44) compared the solubilities of hydrogen in iron, nickel, and cobalt and concluded that they are determined by the crystal structure. The solubility is greatest in the facecentered cubic lattices of y-iron and nickel, less in the body-centered cubic lattice of α -iron, and smallest in the close-packed hexagonal structure of cobalt. In studying a 13 percent manganese steel treated so as to have different amounts of untransformed austenite, Armbruster(37) concluded that manganese per se has little influence on the solubility of hydrogen in iron, the higher solubilities in the 13 percent manganese steel being related to the amount of retained austenite.

Solubility of Hydrogen Isotopes

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The solubility of the hydrogen isotopes also has been studied. Sieverts et al(33) determined the 1-atmosphere isobars for hydrogen and deuterium in two types of iron (ingot iron and iron made by the carbonyl process) for temperatures above 915 F. The solubility of deuterium in iron was found to be a little less than that of hydrogen, and the difference increased with increasing temperature.

Hawkins(45) has compared the solubilities of hydrogen, deuterium, and tritium in an austenitic stainless steel (18.8Cr, 10.66Ni) at temperatures in the range from 570 to 1380 F (300 to 750 C) at pressures of 0.1 to 2.5 mm Hg. The solubilities were given in the following form:

$$\log S = -\frac{A}{T} + B, \qquad (5)$$

where S is the solubility in micromoles per 100 g of steel at 1 mm Hg, and T is degrees Kelvin. The experimental values were as follows:

Isotope	A	B	$^{\Delta H}$, <u>cal/mole</u>
Hydrogen	366,9	1.314	3370
Deuterium	382.2	1.259	3520
Tritium ^(a)	400.2	1.267	3680

(a) T = 96%. H = 4%.

The solubility of 100 percent tritium was calculated and reported also.

Frank, Lee, and Williams(46) calculated the ratio of the solubilities of hydrogen and deuterium over the temperature range from 79 to 187 F (26 to 86 C) from observed permeation and diffusion rates. They used the relationship

$$P_{H}/P_{D} = (D_{H}/D_{D})(S_{H}/S_{D}),$$
 (6)

where P = permeation rate, D = diffusion rate, and S = solubility. D_H/D_D was taken as 1.37 (a value determined experimentally and found to be invariant with temperature). The results follow:

Temperature		Solubility Ratio
С	F	(S _H /S _D)
86	187	6.9
80	176	7.8
79	174	7.4
60	140	4.4
60	140	5.0
52	126	3, 2
52	126	3.4
50	122	4.9
42	108	4.2
34	93	3.6
26	79	3. 3
26	79	3.5

They concluded that, although permeation-rate data are not always highly reliable, there appears to be a trend toward greater difference between the solubilities of hydrogen and deuterium at higher temperatures.

These differences in solubility and diffusivity of the hydrogen isotopes, particularly those obtained by the latter investigators, indicate that tagged, or tracer, experiments would be of limited value for obtaining solubility or rate data for ordinary hydrogen.

Measurements of Solubility at Lower Temperatures (750 F to Room Temperature)

Experimental difficulties have made it impossible to date to obtain as complete an understanding of the solubility of hydrogen in iron and steel for the temperature range between 750 F (400 C) and room temperature as has been obtained for the higher temperatures. This is unfortunate, because the behavior in this lower temperature range is important to problems of hydrogen embrittlement and delayed, brittle failure. In this lower temperature range, the behavior of hydrogen often is changed greatly by variations in impurity content, mechanical and thermal processing, and the method used in preparing the specimen surfaces. Also, the effects of strain and cold work upon the occlusive capacity of metals, including iron, are still controversial.

Effect of Temperature and Pressure

Hill and Johnson⁽¹³⁾ measured equilibrium concentrations of hydrogen in iron at hydrogen pressures up to about 150 atmospheres and temperatures down to 295 F (145 C). Some of the results are shown in Figure 8, where the 100atmosphere isobar for the solubility of hydrogen in one of the heats of iron is shown. The dashed line in the figure was obtained by applying Sieverts' law to the solubility data which Geller and Sun⁽³⁸⁾ reported for temperatures above 750 F (400 C) at a pressure of 1 atmosphere. The agreement between the 100-atmosphere isobar and that given by Sieverts' law is quite good above 735 F (390 C); this indicates that Sieverts' law is followed reasonably well at pressures up to at least 100 atmospheres at these higher temperatures. At temperatures below 735 F (390 C), the experimentally measured solubility deviated markedly from the extrapolation of the high-temperature behavior in the direction of greater solubility. In a detailed analysis of the data, these investigators found this deviation to result chiefly from a change in the heat of solution and only secondarily from deviations from Sieverts' law. This experimentally observed abnormally high solubility of diffusible hydrogen below 735 F (390 C) tends to confirm the hypothesis of hydrogen trapping that Johnson and Hill presented earlier, in Reference (10). They believe that the traps for hydrogen in iron form suddenly on cooling. This theory will be discussed more fully later.



FIGURE 8. SOLUBILITY ISOBARS AT 100 ATM FOR HYDROGEN IN IRON⁽¹³⁾

Hill and Johnson(47, 10, 13) also found that certain, but not all, grades of iron, when treated for periods of several days at high pressures and at temperatures between 840 and 1200 F (450 and 650 C), will absorb considerably more hydrogen than is indicated by the solubility data shown in Figure 8. The hydrogen thus introduced can be extracted in two separate fractions, one of which is evolved in accordance with the usual diffusion coefficient for annealed α -iron*. The second, or extra, fraction is much more difficult to remove and requires extreme extraction conditions. This will be discussed further when the removal of hydrogen from steel is considered.

^{*}For example, see Figure 28 (page 40) and Equations 38 and 43 (pages 40 and 44) to be presented later in further discussion of the work of these investigators.

By using enthalpy and heat-of-solution data for trapped and interstitial hydrogen, Hill and Johnson⁽¹³⁾ were able to estimate the solubility of hydrogen in iron at room temperature. Extrapolating their measurements of solubility at 100 atmospheres and 295 F (145 C) indicated that the 100atmosphere solubility at 77 F (25 C) is between 0.5 and 1.0 ppm, of which less than 0.02 ppm is interstitial hydrogen. Using the directly measured solubility data for hydrogen in iron at elevated temperatures obtained by Armbruster, (37) Darken and Smith⁽⁴⁸⁾ estimated by extrapolation that the solubility at 1 atmosphere and 95 F (35 C) is between 0.9 x 10^{-3} and 0.9 x 10^{-4} ppm.

Bryan and Dodge, (49) in studying the solubility of hydrogen in annealed, high-purity iron, obtained excellent agreement between the 1-atmosphere solubilities calculated from permeability and diffusivity on the one hand and from degassing data on the other, throughout the temperature range from 259 to 1279 F (126 to 693 C).

Effect of Composition

Few data are available to show the effect of steel composition on hydrogen solubility for the lower temperature range. Hudson, Magor, and Stragand⁽⁵⁰⁾ charged six commercial enameling steels, six experimental rimmed steels, and ten commercial steels of quite diverse compositions with hydrogen by acid charging in 2N H₂SO₄ at 100 F (38 C). They obtained hydrogen solubilities, determined by warm extraction under mercury at 320 F (160 C), as follows:

Steel ^(a)	Hydrogen Solubility, cm ³ /100 g of steel
Six commercial enameling steels	s 6 - 40
Six experimental rimmed steels	0-25
USS Man-Ten	31.0
USS Vitrenamel	30.0
Vanadium-alloyed rimmed steel	20.5
Silicon steel	15.0
Rimmed steel	11.9
Bessemer steel	7.5
Aluminum-killed steel	4.5
Ferritic stainless steel (Type 43	2.1
Titanium-alloyed killed steel	1.6
Austenitic stainless steel (Type	302) 1.0

(a) All contained less than 0.20 percent carbon.

It is presumed that the apparent solubility of the austenitic stainless steel was low because of resistance to attack during pickling or because of incomplete extraction. With thermally charged specimens and a higher temperature range, Armbruster(37) showed that austenitic stainless steels had higher solubilities for hydrogen than did ferritic high-chromium steels.

Effect of Cold Work

A number of investigators have shown that for the best available grades of high-purity iron there is little or no effect of cold deformation or of subsequent annealing on the hydrogen solubility or the density. (51,14) However, in the case of SAE 1020 steel, cold work greatly increases the solubility for hydrogen. For example, Darken and Smith⁽⁴⁸⁾ and Hudson, Riedy, and Stragand⁽⁵²⁾ found that cold work greatly increased both the saturation value of solubility and the time required to reach saturation (see Figure 9). Keeler and Davis(51) found that cold work increased the hydrogen occlusion at 480 F (250 C) by a factor of about 100 and lowered the density, the latter effect ascribed to the formation of voids. Hill and Johnson(14) observed that changes in solubility increased in magnitude with increases in carbon content and with increased degree of deformation.



FIGURE 9. EFFECT OF COLD REDUCTION ON RATE OF HYDROGEN ABSORPTION AND ON THE SATURATION VALUE OF HYDROGEN CONTENT; WIRE STOCK IMMERSED IN N H₂SO₄ AT 95 F (35 C); GREEN ROD ¹/₂-IN. DIAMETER⁽⁴⁸⁾

(1 cc NTP per gram = 0.00899 percent H.)

The effects of cold work on the solubility of hydrogen in steel usually are explained by hypothesizing the existence of "traps" in which hydrogen in excess of the equilibrium lattice solubility can be accommodated. (51,14,48) Presumably, the concentration of such traps increases with increased degree of plastic deformation. Apparently, all of the traps cannot be removed by annealing the coldworked steel, since, as will be shown later, such a treatment does not necessarily yield a material with the same characteristics toward hydrogen occlusion as were possessed by the original hotworked steel.

The main conclusions that Keeler and Davis⁽⁵¹⁾ drew regarding the observed effect of cold work in increasing the solubility of hydrogen in SAE 1020 steel were as follows: (1) impurities must be present if the density is to decrease upon cold reduction, (2) the observed hydrogen occlusion is not compatible with the theory of occlusion by the storage of molecular hydrogen in the traps, and (3) two kinds of voids probably result from imperfect plastic flow of ferrite about inclusions or other phases during cold working. Regarding Conclusion (3), they conjectured that hydrogen may be stored in the gaseous form in the larger of the two kinds of voids, which do not disappear on annealing, but it is apparently associated in some other way with narrower imperfections that are eliminated by annealing or recrystallizing. Hill and Johnson⁽¹⁴⁾ proposed that much of the hydrogen contained in cold-worked steels in this lower temperature range is contained in cracks. The apparent heat of solution that they calculated for the cold-worked alloys was similar to the heat of chemisorption of hydrogen on iron powder reported by Emmett and Harkness. (53) Therefore, Hill and Johnson suggested that the traps in cold-worked iron-carbon alloys are cracks, formed by piled-up dislocations, in which the hydrogen is chemisorbed.

Other investigators have reported variations from the behavior just described. For example, Bryan and Dodge(49) found that the solubility of hydrogen in cold-worked high-purity iron (26 percent reduction) was about 40 percent higher than it was in the annealed material. However, the solubility calculated from permeability and diffusivity was about the same for the two materials. This suggested to them that, even in high-purity iron, some hydrogen is trapped in a nondiffusible form, possibly by chemisorption on walls of cracks or as some chemical compound with carbon or other impurities. Duflot(54,55) and Naumann and Carius⁽⁵⁶⁾ working with iron and low-alloy steel, respectively, found first a decrease in the capacity of the metal for hydrogen with increasing deformation, followed by a considerable increase at high deformations. Contrary to these findings, Schuetz and Robertson(57) found that the saturation hydrogen content of steel passed through a maximum with increasing cold deformation. Raczynski and Smia-lowski⁽⁵⁸⁾ observed that cold reduction reduced the occlusive capacity of mild steel. These examples have been cited to show that it is hazardous to try to compare the results of the various investigators because of different experimental conditions or the lack of published information on the conditions used.

Another source of discrepancies is variation in the methods of analysis for hydrogen, which ranged from warm extraction at relatively low temperatures to vacuum fusion. Residual, or "nondiffusible", hydrogen has been widely known in connection with the analysis of steels for hydrogen. It is chiefly responsible for the lack of acceptance in this country of warm extraction as an analytical method for determining hydrogen; in the warmextraction method, only the hydrogen evolved during heating in the alpha range is measured. The residual hydrogen is not all removed from the steel by this technique, so preference has been given to analytical techniques whereby it is possible to recover all the hydrogen, an example being the tinfusion method. Residual, or "nondiffusible", hydrogen will be discussed further in a subsequent section on the removal of hydrogen from steel.

Smialowski⁽²⁴⁾ pointed out that experiments with steels quite similar in composition but which differ in the distribution of impurities often show great differences in the occlusive capacity for hydrogen. He is of the opinion that this effect of impurity distribution is probably of greater importance than the amount of cold work.

Podgurski⁽⁵⁹⁾ studied the anomalously high solubility reported for hydrogen in cold-worked steel at temperatures somewhat above 390 F (200 C). He attributed this to methane formed by a reaction between hydrogen and cementite; he quantitatively isolated this methane from steel. Furthermore, the calculated volume needed to accommodate the amount of methane found was approximately equal to the volume corresponding to the measured density change resulting from the cold work. Thus, he concluded that the methane formed is accommodated in microvoids in the metal. However, isotope experiments ruled out the possibility that methane is produced in steel at room temperature by acid charging.

Effect of Annealing After Cold Work

Although annealing a steel after cold deformation may eliminate the density loss and reduce the large occlusive capacity, it does not necessarily yield a material with the same characteristics toward hydrogen occlusion as those possessed by the original hot-worked steel (see Figure 10). Apparently, all of the traps for hydrogen cannot always be removed by annealing the cold-worked steel. Keeler and Davis(51) found that the traps in cold-worked SAE 1020 steel persisted to temperatures up to about 750 F (400 C). This is the same temperature as that below which anomalously large solubilities are found for hot-worked or annealed materials.

Hudson, Riedy, and Stragand⁽⁵²⁾ studied the influence of cold work on hydrogen absorption rate into, solubility level in, and desorption rate from low-carbon steel sheet. In one part of the study, samples of a commercial steel were obtained in



FIGURE 10. THE EFFECT OF COLD DRAWING (75.9%) AND OF SUBSEQUENT AN-NEAL [64 HOURS AT 1470 F (800 C)] ON HYDROGEN ABSORPTION AND THE SATURATION VALUE OF HYDRO-GEN CONTENT; WIRE STOCK IMMER-SED IN N H₂SO₄ AT 95 F (35 C)⁽⁴⁸⁾

(1 cc NTP per gram = 0.00899 percent H.)

the as-cold-reduced, as-continuously-annealed, and as-box-annealed conditions from the same original coil. The box-annealing operation involved 4 hours at a temperature of 1200 F, whereas the continuous strip anneal involved a period of less than 1 minute at this temperature. The effects of these treatments on the time for hydrogen to reach the saturation level and on the saturation concentration were as follows:

Condition	Time to Reach Saturation Level, minutes	Hydrogen Saturation Concentration, cc/100 g steel	
Cold reduced	>50	>78	
Continuously annealed	10	12	
Box annealed	19	5	

Note that the longer annealing time associated with the box anneal reduced the saturation level for subsequent hydrogen charging considerably. Darken and Smith⁽⁴⁸⁾ also showed that increased annealing time (and increased temperature too) lowered the hydrogen saturation concentration reached during subsequent pickling of the steel.

In a continuation of the work reported in Reference (52), Hudson et $al^{(60)}$ found that the effect of cold reduction on hydrogen solubility in mild steel persisted through subsequent heat treatment and later cold reduction as well. This was not true, though, for hydrogen permeability.

Effect of Surface Conditions

As one might expect, the surface conditions of the steel have no effect on the hydrogen saturation level, but instead, they affect the rate of entry. This is illustrated by the work of Hudson, Magor, and Stragand⁽⁵⁰⁾ in which they found that, for a given steel, the rate of hydrogen absorption during acid pickling was greater for a heavily sandblasted surface than for a lightly sandblasted specimen, although the saturation level was not affected by the difference in surface conditions. This behavior, shown in Figure 11, illustrates the need for obtaining reproducible surface conditions when permeabilities are to be measured and compared.



FIGURE 11. INFLUENCE OF SURFACE CONDITION ON HYDROGEN CHARGING RATE⁽⁵⁰⁾

Estimates of Equilibrium Hydrogen Pressure in Steel

As was shown in preceding sections, when a piece of steel cools, the hydrogen solubility changes. The solubility of atomic hydrogen in steel under a given pressure of molecular hydrogen decreases with decreasing temperature. As the equilibrium solubility is reduced, the hydrogen tends to escape from surfaces. Microshrinkage cavities, gas holes, and other voids and discontinuities in the steel provide internal surfaces, and atomic hydrogen in the surrounding steel will precipitate in them as molecular hydrogen cannot diffuse in steel.

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As the steel adjacent to the void becomes lower in hydrogen, a concentration gradient is created which results in continued flow of hydrogen toward the void and continued precipitation there. At room temperature, atomic hydrogen can diffuse and precipitate against considerable pressure of molecular hydrogen. However, there is disagreement as to the magnitude of these pressures. For years, it was considered that very high equilibrium pressures prevailed. It is well known that iron or steel is readily blistered at high temperatures by molecular hydrogen that forms in voids. The very fact that a blister forms shows that the pressure is greater than the yield strength of the steel. Also, analyses of gas extracted from blisters have shown the gas to be hydrogen. In cases where the steel is attacked by hydrogen and becomes decarburized, methane (CH_4) is found. The attack of steel by hydrogen at elevated temperatures and high hydrogen pressures is the subject of a separate DMIC report (DMIC Report No. 202).

Many investigations have shown that hydrogencontaining steel, on cooling to ordinary temperatures, is highly amenable to supersaturation with hydrogen, and consequently, to internal accumulations of hydrogen under high pressure. Pickling and cathodic electrolysis also are means of producing a high degree of supersaturation, because these processes create concentrations of atomic hydrogen on the surface of the steel that are equal to, or greater than, the concentrations occurring in hydrogen gas at high temperatures. On a number of occasions, the present authors have observed blistering of steel during cathodic charging with hydrogen at room temperature, and other instances of such behavior have been reported in the technical literature. This is visual evidence that high pressures are produced at room temperature.

Taking into consideration the fact that hydrogen does not behave as a perfect, or ideal, gas at high pressures, Carney, Chipman, and Grant⁽⁶¹⁾ calculated the equilibrium hydrogen pressures shown in Table 4. For a hydrogen content in the steel of 0.00050 percent, or 5 ppm (an amount commonly found in steel not deliberately charged with hydrogen), an equilibrium hydrogen pressure of 184,000 psi was indicated at room temperature. Luckemeyer-Hasse and Schenck⁽³⁵⁾ used the squareroot-of-the-pressure law to calculate equilibrium pressures of molecular hydrogen necessary at an internal surface to prevent atoms of the gas dissolved at some higher temperature from precipitating during cooling. They, too, showed that normal hydrogen contents in steel (as determined by analysis) might be expected to provide very high stresses in voids in the steel. For example, their results indicated that, with molten steel with hydrogen contents in the order of 0.0010 percent (10 ppm) and above, pressures exceeding the tensile strength of the casting are reached before the steel has cooled to room temperature.

TABLE 4. EQUILIBRIUM PRESSURES OF HYDROGEN IN STEEL(61)

Hydrogen in Steel, %	Fugacity, atm	Pressure of Hydrogen, atm	Pressure of Hydrogen, psi
	At Room Tem	perature	
0.00100	30,800,000	14,500	218,500
0.00050	7,720,000	12,250	184,000
0.00020	1,240,000	9,325	140,000
0.00009	250,000	6,850	103,000
	At 390 F (20	00 C)	
0.00100	2,580	1,430	21,450
0.00050	647	520	7,800
0.00020	103	99.4	1,490
0.00009	20.9	20.9	314
	At 750 F (40	0 C)	
0.00100	820	665	9,970
0.00050	205	193	2,900
0.00020	32.6	32.6	409
0.00009	6.6	6.6	99

Experiments with hollow cathodes have produced pressures of 200 to 300 atmospheres (about 3,000 to 4,500 psi), (62,63) but these fall short of being equilibrium values, since the experiments were discontinued when the limits of the pressure-measuring equipment were reached. Smialowski and co-workers have used a method for estimating the pressure that is based on the measured expansion of iron when charged cathodically in an acid electrolyte with cathodic poison. They believe that the expansion is caused by molecular hydrogen that is trapped in cavities formed at nonmetallic inclusions, microporosity, and lattice defects.⁽²⁴⁾ The magnitude of the expansion observed suggests that the pressures developed exceed 10⁵ atmospheres (1,470,000 psi), possibly reaching 107 atmospheres (147,000,000 psi).

According to Evans, (64) the pressures attained experimentally are small compared with those theoretically possible. From studies of hydrogen overvoltage in cathodic charging, one might conclude that the pressure might be of the order of 10^7 atmospheres. However, this assumes that the electrochemical laws continue to apply at high levels of overvoltage and that there is no leakage of molecular hydrogen - conditions that are not likely to be met. More refined calculations by Ubbelohde and co-workers(65, 66) suggest that pressures of about 6.6 x 10^{18} atmospheres theoretically ought to be obtained.

Experiments have shown that before such high pressures could be developed, passageways would have been opened that would permit the escape of molecular hydrogen, and the pressure would no longer rise. Both $Bridgman^{(67,68)}$ and Poulter and Uffelman⁽⁶⁹⁾ conducted experiments which demonstrated that molecular hydrogen under high pressure can open pathways in steel through which molecular hydrogen can diffuse. Bridgman

found that a hollow steel cylinder containing hydrogen gas under pressure allowed the gas to escape with explosive violence when the pressure reached 128,000 psi, but without visibly damaging the cylinder. Poulter and Uffelman subjected a cylinder, which withstood 20,000 atmospheres (294.000 psi) oil pressure, to hydrogen gas pressure of 4,000 atmospheres (59,000 psi). After only 5 minutes, there was a rapid evolution of hydrogen through the cylinder wall. Upon filling with oil at only 4,000 atmospheres (59,000 psi), oil passed through the wall and the pressure dropped to 500 atmospheres (7,350 psi) in 2 hours. Examination under a low-power microscope revealed no flaws. However, when filled with hydrogen at 100 psi and immersed in water, small bubbles soon appeared over the entire outer surface.

At the other extreme, Hill and Johnson(13)purportedly showed that the pressure of molecular hydrogen in equilibrium with an atomic hydrogen concentration in the steel of 5 ppm at room temperature must be several orders of magnitude lower than the value usually extrapolated from above 750 F (400 C). By their method of calculation, these hydrogen pressures are less than 1000 atmospheres, that is, below 14,700 psi. Thus, they suggest that the common theory that hydrogen embrittlement of steel results from extremely high equilibrium pressures of molecular hydrogen is incorrect. Of course, hydrogen contents much higher than 5 ppm can be introduced into steel by suitable charging techniques, and these would correspond to higher equilibrium pressures than 1000 atmospheres, which might account for the blistering observed.

PERMEATION

Introduction

Sometimes the terms "permeation" and "diffusion" have been used more or less interchangeably in describing the penetration of gases through metal walls. However, a number of writers concur in the belief that it is better to make the following distinction between the two. The overall process from bulk gas phase on one side of the metal wall, through the metal, to the bulk gas phase on the other side is best termed permeation. This consists of interactions at both the entry and the exit surfaces of the metal, as well as the actual transport through the metal. The term diffusion strictly applies only to transport through the metal. In the usual experiment with hydrogen, the permeation, rather than the diffusion, is measured, because one is dealing with the overall driving force of gas pressure.

Since the presence of hydrogen (sometimes as little as 1 ppm) may have a significant effect on the properties of steel and on the tendency toward delayed cracking of steel under sustained load, knowledge of the processes by which it enters and leaves steel is desirable. For convenience, Darken and Smith⁽⁴⁸⁾ divided the processes into two classes which they termed "stationary" and "nonstationary".

The stationary state may be associated with the passage of hydrogen through a diaphragm. If conditions are maintained constant, the rate of passage of hydrogen through the diaphragm usually becomes constant (stationary) after sufficient time. This steady-state condition is described by the permeability constant, or permeability.

The nonstationary state is associated with the rate at which a solid steel object absorbs hydrogen from a suitable environment or loses hydrogen after it has been removed from that environment. This nonstationary diffusion phenomenon usually is described by the boundary conditions and the diffusion constant, or diffusivity.

For linear, steady-state diffusion with a concentration gradient as the driving force, the coefficient of diffusivity, D, may be defined by the equation

$$J = -D \frac{(c_1 - c_2)}{\Delta x} , \qquad (7)$$

while permeability, P, is usually defined by

$$J = \frac{-P}{\Delta x} , \qquad (8)$$

where J is the diffusion current per unit of area and $(c_1-c_2)/\Delta x$ is the concentration gradient in the metal wall of thickness Δx . The permeability depends upon the gas-pressure difference on the two sides of the wall, so a permeation coefficient could be defined based on a pressure gradient. However, the usual values of P have been obtained with a pressure difference of 1 atmosphere. Inasmuch as it has been established that the concentration of hydrogen in steel is proportional to the square root of the pressure, the permeability, P, also might be defined by the equation

$$J = -P \frac{\sqrt{p_1} - \sqrt{p_2}}{\Delta x} .$$
 (9)

Permeability may be expected to depend on temperature, the nature of the metal, the boundary conditions, and the thickness of the diaphragm; there are indications that it may also depend on the nature of the environment surrounding the exit surface. For the permeation of hydrogen through steel, the boundary conditions include the nature of the steel surface and such things as the type and concentration of the acid in contact with one side of the diaphragm in pickling, the partial pressure of hydrogen in the gas used in pressure charging, and the nature of the electrolyte and the current density in cathodic charging from one surface. The presence or absence of a poison in pickling or in cathodic charging is another factor which is very important. Thus, many variables affect the rate at which hydrogen permeates metals, and, since apparently conflicting results have been

obtained, the process is not completely understood. For example, consider the effect of pressure. It has been established reasonably well that the permeation rate of hydrogen through steel is proportional to the square root of the pressure (except at very low pressures). This indicates that it is only atomic hydrogen that penetrates steel and permeates through it. However, according to Dodge (70) with many steels the permeation rate becomes constant when a hydrogen pressure of about 100 atmospheres is reached, and no further increase in permeation rate is obtained with further increases in hydrogen pressure. Apparently, as yet there is no satisfactory explanation of this behavior.

 $Dodge^{(70)}$ has given the following equation as being the one usually accepted as expressing the effect of pressure and temperature on permeation:

$$J = \frac{K}{\Delta x} (p_1^{\frac{1}{2}} - p_2^{\frac{1}{2}}) e^{-E/(RT)}, \quad (10)$$

where J = rate of permeation per unit area

- K = a constant that depends on the metal and probably on several factors related to the state of the metal
- Ax = thickness of metal membrane through which gas is permeating
- p₂ = pressure on low-pressure side
- E = activation energy
- R = gas constant
- T = absolute temperature.

Although this law of pressure and temperature dependence is supposedly general, the limits within which it applies and the effect of variables on K are not well established.

A number of investigators have measured the amount of hydrogen permeating through a thin tube and agreed that the following equation holds:

$$P(or Z) = K p^{\frac{1}{2}} e^{-B/T},$$
 (11)

where p is the pressure, T is the absolute temperature, and K and B are constants. For the pressure in mm and temperature in degrees Kelvin, Smithells and Ransley⁽⁷¹⁾ gave the following values for the constants for iron:

$$K = 1.63 \times 10^{-3}$$
 and $B = 4800$.

Borelius and Lindblom(72) obtained

$$K = 1.60 \times 10^{-3}$$
 and $B = 4700$

for high-purity iron. The effect of pressure on hydrogen permeation will be discussed in more

detail in the section dealing with hydrogen entry. Because the greatest effect of temperature on permeation is through its effect on diffusion, further consideration will be given to temperature in the section on diffusion.

The permeability, P, is related to the solubility, S, and the diffusivity, D, as follows:

$$P = S \cdot D$$
. (12)

A number of investigators have used this relationship to calculate the diffusivity of hydrogen in iron or steel from permeability data obtained experimentally. However, Bryan and Dodge(49) indicated that this relationship holds only for the case where the diffusion through the bulk metal is the controlling rate, and surface resistances to permeation are negligible. It will be shown later that these three properties are mutually consistent for hydrogen permeating through iron or steel at temperatures above about 750 F (400 C).

Many investigators have studied the rate at which hydrogen permeates metals, and a large number of these studies have been performed on iron and steel. For that matter, a permeation experiment is the usual method used to study the movement of hydrogen in steel. However, because of the great number of factors that influence permeation, the analysis of the results is complicated. Actually, the person dealing with hydrogen in steel usually is interested in hydrogen entry, hydrogen diffusion through the bulk solid, and the removal of hydrogen. Although these processes are all part of permeation, it will be more convenient for the purposes of this report to consider them separately.

Hydrogen Entry

Hydrogen enters steel only when it is presented to the steel surface as atomic hydrogen, as has been discussed previously in the section on solubility. Those ways in which hydrogen enters steel subsequent to its manufacture are of special interest to the defense contractor who is concerned about hydrogen embrittlement and delayed brittle failure. These are corrosion, electrolytic processes, and dissociation of hydrogen gas at elevated temperatures.

Entry by Corrosion Processes

Atomic hydrogen can be liberated at the surface of iron or steel as a result of direct attack by corrosive media or galvanic corrosion. Important examples of the former are acid attack as in acid pickling, attack by hydrogen sulfide, and corrosion by water such as has resulted in hydrogen cracking of pressure vessels being pressure tested with water. Galvanic corrosion is the accelerated electrochemical corrosion that occurs when two dissimilar metals are joined electrically and are in the same corroding medium or electrolyte. This type of

corrosion usually produces a higher rate of corrosion on the less noble of the two metals and tends to protect the more noble metal; hydrogen may be evolved on the latter. Thus, hydrogen may be evolved on iron or steel when it is coupled with a more active metal, such as magnesium, aluminum, or cadmium.

In any diffusion or permeation problem, the boundary conditions are of great importance. Therefore, it is desirable to have some measure of the driving force of hydrogen; this usually is given in terms of pressure or concentration. When atomic hydrogen is liberated at the surface of steel by chemical attack (for example, by immersion in a suitable acid solution), no actual gage pressure (above atmospheric) of hydrogen is developed on the steel surface. If the pressure concept is adopted for this situation, it must be considered to be a virtual pressure. For an ideal gas, the driving force may be considered to be the fugacity, a thermodynamic quantity which may be regarded as a potential or virtual pressure. Three independent methods of calculating fugacity for hydrogen on iron immersed in $1N H_2SO_4$ all give values that lie between 10^6 and 10^8 atmospheres, (48) pressures far higher than the bursting strength of steel. As Sims⁽⁷³⁾ has stated, such reactions as those of water vapor, pickling in acid, or cathodic precipitation may create an atmosphere at the steel surface, which, in terms of hydrogen-atom concentration, may be equivalent to thousands of atmospheres of hydrogen gas at room temperature or to lowpressure hydrogen at very high temperatures. In this way, the high concentrations of hydrogen that may be obtained in steel by pickling or other chemical attack or by cathodic charging may be accounted for. The fugacity varies with the electrical potential developed in the acid, and this depends on which acid is used, and on its concentration and temperature. (48)

Acid Pickling. It is necessary to remove mill scale or rust from steel before applying a metallic coating, and such removal is advantageous before applying paint. Although this can be done by mechanical means, such as grit blasting or grinding, it often is carried out by pickling in acid solutions. Part of the action of pickling consists of an undermining of the scale which results, in part, from attack of the steel itself by the acid. Electrochemically, iron is a less "noble" element than hydrogen, and thus, it replaces hydrogen from solutions of its ions (acids). Thus, iron is corroded in acid solutions with a pH less than about 4. When a piece of iron or steel is placed in a solution of a nonoxidizing acid with which it reacts, as in pickling, there are local anodes and cathodes acting, and atomic hydrogen is released at the local cathodes on the steel surface, according to the reaction

 $2H^+ + Fe \longrightarrow Fe^{++} + 2H.$ (13)

A portion of this liberated nascent hydrogen enters into the steel. The portion that does not enter the steel almost immediately, then combines to form molecular hydrogen which is not soluble in metals. The fraction of the hydrogen atoms that recombine and thus escape from the steel surface without entering is greatly affected by inhibitors or by promoters (promoters are commonly called cathodic poisons). When a piece of steel which has been pickled is removed from the acid solution and held in air, the hydrogen which it contains tends to effuse slowly out of the steel.

Entry of hydrogen into the steel is highly undesirable because it may produce blisters in the steel, or hydrogen cracking may occur, either during pickling or afterwards when the part is placed under load. Acid pickling may introduce more hydrogen into steel than is introduced by electroplating operations.

One method used to reduce destruction of valuable metal, minimize surface roughening, and reduce unnecessary consumption of acid is to make the workpiece a cathode by means of an external potential source, rather than merely dipping it in the acid. However, problems related to the absorption of atomic hydrogen by the steel are likely to become more severe when pickling is performed cathodically. Blistering and cracking become even more severe if arsenic or selenium, which are common impurities in acids, are present during cathodic pickling. However, by adding a trace of tin salt to the acid, it is possible to realize the advantages of cathodic pickling while minimizing the disadvantages. A very thin layer of metallic tin is deposited on any portion of the metal which becomes bare of scale. This layer hinders evolution of hydrogen because of the high overvoltage of tin, and it also imposes a barrier to retard the entry of atomic hydrogen into the steel.

One way to reduce the evolution of hydrogen resulting from attack of the metal is to make the steel the anode during acid pickling (anodic pickling). It should effectively prevent the evolution of hydrogen on the steel and, therefore, its entry into steel. However, sometimes anodic pickling increases both the attack on the metal and the roughening, and it may increase acid consumption. Therefore, this procedure is used most frequently for material from which the scale previously has been removed mechanically, the object being to produce a surface suitable for electropolishing.

The method most frequently used to reduce the problems related to metal attack and hydrogen liberation at the same time is to add to the sulfuric or hydrochloric acid pickling bath a substance that will restrain the attack on the metal while still permitting the scale to be undermined. Such substances usually are called "inhibitors". However, Evans(64) has pointed out that the corrosion rate is not brought down to so low a level as to justify use of this term, the corrosion rate being tolerated only because the time of immersion in the acid is

comparatively short. He prefers use of the term "restrainer". Restrainers are commonly used in ordinary acid pickling baths without an external applied electric current. However, a combination of restrainer and cathodic current can be a very economical procedure. With this arrangement, the current required is much lower than in the absence of a restrainer, and the restrainer concentration required is far below that needed in the absence of current. It is claimed that the combination avoids the chief difficulties encountered with each method, that is, the problem of hydrogen embrittlement and delayed cracking associated with cathodic treatment at high current densities, and the loss of time associated with the use of restrainers, some of which appreciably retard the removal of scale. The current density can be reduced to 1/100 that needed when no restrainer is used, and the restrainer concentration can be reduced to 1/10 that needed when no current is employed.(74)

Evans, (64) in describing the use of restrainers, pointed out that if the only objective in using a restrainer is to cut down the destruction of metal and the consumption of acid, any substance can be used which obstructs any one of the essential steps of the corrosion process. It would make little difference whether the step obstructed is the anodic reaction or one stage of the cathodic reaction. A large number of substances are known to appreciably diminish the corrosion rate, including such cheap materials as glue, size, gelatin, and molasses. However, in practice a restrainer must meet other requirements. Reducing the total corrosion, that is, the total hydrogen production, does not suffice. It is particularly necessary to reduce that fraction of the hydrogen which enters the metal and causes either blistering or cracking. A restrainer which poisoned the second stage of the cathodic reaction without interfering with the first stage of the cathodic reaction or with the anodic reaction would be harmful in many instances, because it would cause a concentration of atomic hydrogen to be built up and thus actually would promote a diffusion of hydrogen into the metal. For example, glue additions greatly reduce the amount of hydrogen gas (molecular form) evolved from the pickling bath but have little effect on the risk of cracking.

Formerly, many commercial restrainers were used which did not meet all these requirements. Zapffe and Haslem⁽⁷⁵⁾ tested 14 of them and found that on stainless steel 11 of them actually increased hydrogen embrittlement, and none of them reduced it appreciably. On mild steel, some of the restrainers produced considerable improvement. Some recent work on the effect of restrainers on the evolution of hydrogen during pickling is discussed in Reference (76).

Many of the restrainers used today are organic substances that contain nitrogen and sulfur. It is believed that these polar atoms serve to attach

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otherwise some essential step in the corrosion process would occur. The proprietary restrainers now available contain such compounds as diorthotolyl-thio-urea or dihydro-thio-ortho-toluidine. Substances with nitrogen in the ring, such as quinoline and its derivatives, also are effective restrainers. Generally, the efficiency of the restrainer increases as the length and number of side chains attached to the polar atom increase. However, some small molecules, such as formaldehyde, can act as restrainers. Most of the present effective restrainers interfere mainly with the anodic reaction, although the cathodic reaction usually is retarded somewhat also.

A list of pickling inhibitors, including information on the amounts required, may be found in Reference (77).

The use of restrainers, or inhibitors, has been discussed for the person who desires to use pickling in processing his steel and still reduce the amount of hydrogen introduced into the steel. However, for high-strength steels, which are especially susceptible to hydrogen-induced delayed brittle failures, it is best to avoid pickling and all cathodic cleaning processes. For these materials, scale is best removed mechanically, and only alkaline or anodic cleaning processes should be used.

For the person in the laboratory who is studying the effects of hydrogen on steel, it may be desirable to introduce hydrogen into steel intentionally. Sometimes this is done by acid pickling, though more frequently it is done electrolytically by cathodic charging. In either instance, the entry of hydrogen is greatly facilitated by the presence of substances known as cathodic poisons. Such substances as sulfur, phosphorus, arsenic, H2S, PH3, AsH3, and low-oxidixed compounds of sulfur, phosphorus, and arsenic greatly enhance hydrogen penetration into metals. Also, it is believed that sulfur and phosphorus contained in iron and steel act in a manner similar to that of their compounds added to the acid solution, because, with acid attack of sulfide and phosphide inclusions present at the surface of the sample, H₂S and PH₃ are formed. (78) This subject will be discussed further in the section dealing with entry of hydrogen by cathodic charging, since most of the investigations have utilized this method of hydrogen charging.

Darken and Smith⁽⁴⁸⁾ carried out a comprehensive study of the behavior of hydrogen in steel during and after immersion in acid. They studied the rate of absorption, saturation solubility, and permeation of hydrogen in steel. The method of investigation consisted of exposing one side of a thin steel diaphragm to an acid solution and collecting and measuring the hydrogen which diffused through and passed out the other side. The solubility of hydrogen was determined by both warm

extraction and vacuum extraction at temperatures up to 1470 F (800 C). As one might expect, they found that the amount of hydrogen passing through a steel disc in contact with acid depended on the pH of the acid, as is illustrated in Figure 12. The amount of hydrogen absorbed by the steel also depended on pH, as is shown in Figure 13. In experiments with cylinders of various diameters immersed in a buffered acid solution of pH = 3, 2, the rate at which hydrogen was absorbed was found to be related to the dimensions in the manner to be expected from the usual diffusion laws. That is, a plot of the square root of time for saturation versus specimen diameter gave a straight line. Cold work markedly increased the absorption rate, the time to reach saturation, and the saturation concentration or amount of absorbed hydrogen. For a given material, the amount absorbed increased linearly with the square root of time until a limiting amount was absorbed. After this saturation value was reached, no further hydrogen was absorbed for longer immersion times. The saturation value was very small for hot-rolled steel, but increased markedly with cold work. The effect on absorption of annealing after cold work was less than had been anticipated, inasmuch as the annealed material did not behave toward hydrogen at all like the original hot-rolled material (see Figure 10).



FIGURE 12. AMOUNT OF HYDROGEN (1 cm on scale = 4.27 cc) PASSING THROUGH AN SAE 1020 STEEL DISC IN CONTACT WITH AN ACID SOLUTION PLOTTED AGAINST TIME IN DAYS(48)

		Disc	2
Curve	Solution	Original Thickness, mm	Diameter In Contact, mm
A	pH 6.25	0.965	27.6
в	5.47	0.965	29.0
С	4.33	0.960	27.5
D	10N H2SO4	0.957	27.0
	-		



FIGURE 13. AMOUNT OF HYDROGEN ABSORBED BY SIMILAR SPECIMENS IMMERSED IN A SOLUTION OF pH (A) 7.92; (B) 7.09; (C) 5.43; (D) 3.14; (E) 1.38, PLOTTED AGAINST SQUARE ROOT OF TIME OF IMMERSION⁽⁴⁸⁾

(1 cc NTP per gram = 0.00899 percent H.)

Nitric acid, an oxidizing acid, has been regarded as causing no hydrogenation. (79,64) Most of the base metals that liberate hydrogen from hydrochloric acid are attacked also by nitric acid. However, the hydrogen atoms react with nitric acid to form salts of ammonia or hydroxylamine (NH2OH); these compounds are rich in hydrogen. The more noble metals that cannot liberate hydrogen from hydrochloric acid or sulfuric acid are unlikely to produce hydrogen or hydrogen-rich products such as ammonia salts. Instead, they produce compounds rich in oxygen, such as NO2, HNO2, or NO. Iron shares the properties of both groups and produces both ammonium salts and nitrous acid, besides the gases NO2 and NO. Evans⁽⁶⁴⁾ pointed out that it may seem remarkable that a simple dip in an oxidizing acid bath which would pickle a surface without introducing hydrogen is not more widely adopted as a simple means of avoiding hydrogen troubles without demanding an external potential. However, he states that practically every oxidizing acid or acidic mixture introduces some new objection. Nitric acid, for instance, is unpleasant owing to the fumes evolved and the action on the skin. It was used in one situation where spring clips were found to become

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embrittled on pickling in hydrochloric acid or sulfuric acid followed by electrotinning; the problem was solved by using nitric acid for pickling and applying the tin by hot dipping.⁽⁸⁰⁾ However, Hudson, Norris, and Stragand⁽⁸¹⁾ found that, at least under some conditions, treatment of steel with nitric acid can introduce hydrogen into the steel. This is shown in the following tabulation:

	Concen-		Hydrogen Content,(a) cc/100 g			
Solution	tration, M	pН	Warm Extraction	Vacuum Fusion		
Uncharged			0.1	1.3		
H ₂ O, deoxy- genated		6.6	0.1			
HNO ₃	0.1	1.3	3.6			
HNO ₃	15.6	<0.0	1.0			

(a) Content after treatment by immersion in the indicated solution at 100 F (38 C) for 4 hours.

When fully charged to saturation by immersion in $2N H_2SO_4$ at 100 F (38 C) for 24 hours, average hydrogen contents of 12.7 cc/100 g and 16.0 cc/ 100 g were obtained by warm extraction and vacuum fusion, respectively.

Smialowski⁽⁷⁹⁾ has devoted Chapter 15 to hydrogen in pickling of steel, and he gives $6\frac{1}{2}$ pages of references on the subject.

Other Corrosion Processes, Another source of hydrogen in steel arises from the corrosion of steel in aqueous environments. Time-dependent delayed failures have been observed in rocketmotor cases when tested at constant pressure and exposed to aqueous environments.(82,83) In a series of experiments with small-size pressure vessels, it was proved that the water used in hydrostatic testing was the cause of the delayed, brittle failures, and that these failures were induced by hydrogen. This type of failure is not encountered when oil is used as the pressurizing medium, or when water is prevented from contacting the steel by use of a primer on the steel. However, when water contacts the bare metal, the burst strength is lowered (by 130,000 psi in the experiments mentioned above). The water serves as a medium for localized galvanic action which releases the essential atomic hydrogen. Pressure testing with oil is an effective preventative measure against embrittlement, provided that the pressure-vessel steel has not already been loaded with hydrogen prior to hydrostatic testing. (83)

Hydrogen will permeate steel at room temperature because of corrosion that occurs as a result of exposure to air and liquid water. The corrosion reaction occurring on bare steel under these conditions may be written

 $Fe + H_2O \longrightarrow FeO + 2H.$ (14)

Experiments with heavy water (D_2O) , in which permeation was detected with a mass spectrometer, showed that the source of the hydrogen is the water that participates in the initial corrosion reaction (Equation 14), rather than the hydrogen present in the steel prior to corrosion. (84) When air was removed from the environment surrounding the entry surface, only D_2O being left, the rate of permeation was reduced markedly; this shows that oxygen from the air is involved in the process.

Using a metal radio tube, having an SAE 1010 carbon steel shell, as an ionization gage, Norton⁽⁸⁵⁾ measured the permeation of hydrogen through the steel shell into the interior. The hydrogen was generated by corrosion that occurred when the tube was dipped into water. A bare, sandblasted, steel surface admitted hydrogen at the rate of 0.13 micron/hour with the shell in water at 77 F (25 C). With a protective paint on the tube, the rusting reaction was essentially stopped, and very little hydrogen entered the steel (0.008 micron/ hour). When 0.1 percent sodium chromate, a corrosion inhibitor, was added to the water, entry of hydrogen into even the bare steel was virtually stopped, the entry rate being <0.001 micron/hour. Of the various coatings examined, one coating, zinc (a more active metal than steel), gave an increased rate of hydrogen entry.

With this technique, many hours of immersion were required for the experiments in the case of water rusting the shell. Also, it was a passive system, that is, the tube was turned on only after the diffusion had occurred. McCaldin⁽⁸⁶⁾ developed a method, using a different tube, to continuously determine hydrogen permeating through the shell of the tube. He was able to use this device to measure hydrogen permeation during electroplating, sandblasting, abrading, and chemical clean-ing treatments. Lawrence⁽⁸⁷⁾ used this same type of equipment to continuously monitor hydrogen sorption during the entire electroplating procedure, during stress corrosion, and during exposure to gaseous and aqueous media. As a result of his experiments, the low-embrittlement cadmiumplating process for high-strength steel used by his company was altered materially. It was found that the rinses following pickling or plating were a source of considerable extra hydrogen in the plated parts. The new process now requires that the parts be rinsed in overflow tanks, and that the rinse water be changed at least once each hour to avoid damaging acid buildup. Because it was observed that vapor-honing operations resulted in sufficient hydrogen pickup to cause embrittlement, dry honing, which gives less hydrogen, was recommended.

Exposure to water vapor also introduces hydrogen into steel. Exposure to superheated steam has resulted in measurable hydrogen permeation in both a plain-carbon ferritic steel and austenitic Type 304 stainless steel. (88) The results for ST37 steel (plain carbon) exposed to steam

at 1 atmosphere and temperatures in the range from 350 to 575 F are shown in the top part of Table 6. For comparison, these investigators also listed the permeation rate of hydrogen from water through steel which they computed from Norton's data⁽⁸⁵⁾. By extrapolating their data for steam to temperatures at which Norton determined the permeation rate for water, the results indicated that the permeability of hydrogen from water does not differ greatly from the permeability of hydrogen from steam of the same temperature and pressure, though the density of water is in the order of 2000 times the density of steam. Grinding or abrading steel causes hydrogen pickup, apparently from water vapor in the air or from the cutting fluid when the latter is used. (89,90) A method has been devised to introduce hydrogen into steel for diffusion studies at room temperature and slightly elevated temperatures that consists of abrading steel under water under controlled conditions. (91)

TABLE 6. PERMEATION RATE OF HYDROGEN FROM WATER OR STEAM THROUGH ST37 PLAIN-CARBON STEEL 0.1 mm THICK⁽⁸⁸⁾

Temperature		Environ-		Permeation Rate.	
К	F	ment	Investigators	cm ³ (STP)/(cm ² -sec	
575	575	Steam	Boeschoten et a	al 274 x 10 ⁻⁸	
525	485	St am	Ditto	10×10^{-8}	
450	350	Steam		1×10^{-8}	
373	212	Water	Norton(85)	9×10^{-10}	
298	77	Water	Ditto	1.5×10^{-10}	

Steel ball bearings have been found to pick up hydrogen from a hydrocarbon lubricant. (92) Another application of the radio-tube ionization gage involved a study of the problem of hydrogen generated by paint strippers which often are used on high-strength steel parts. (87) The stripper being used gave a hydrogen input equivalent to that which might have been experienced if the part had been soaked in a H_2SO_4 solution of pH 2 at room temperature. Since it was the practice to allow the stripper to remain on some parts for as long as 24 hours, this aerospace company now requires that all high-strength steel parts exposed to this particular stripper be baked to drive out absorbed hydrogen before further processing.

It is rather generally agreed that the phenomenon of sulfide-corrosion cracking of steel is the result of the penetration of atomic hydrogen into the steel. (57,93,94) That is, sulfide-corrosion cracking is identical with hydrogen cracking. The only part played by corrosion in this phenomenon seems to be to supply the nascent hydrogen to the surface of the steel. Since the sulfide ion is one of the most effective poisons of the hydrogenrecombination reaction, its presence serves not only to corrode the steel and supply the atomic hydrogen, but to make it possible for much of this atomic hydrogen to enter the steel. Inasmuch as only very small quantities of hydrogen embrittle steel and serve to make it susceptible to delayed brittle failure, enough hydrogen to embrittle the steel may be introduced at an insignificant corrosion rate. Both martensitic and ferritic structures fail when the strength level of the steel, the magnitude of the applied stress, and the hydrogen content introduced by sulfide corrosion are high enough, but austenite has not cracked under similar conditions - just as with failures where hydrogen is introduced by cathodic charging or thermal charging, (57,93,94) For a given strength level, there appears to be a minimum stress necessary to produce failure. (93) A number of investigators have concurred in the finding that cold work increases the susceptibility of ferritic steel to sulfide-corrosion cracking and decreases the time to fracture. (57,93,94) In recent years, Bastien and co-workers and Herzog have published the results of a large number of studies of sulfide-corrosion cracking. A number of other investigators have studied the hydrogen embrittlement and delayed brittle fracture of steel resulting from exposure to H2S; some of these studies will be found in References (95) to (101).

Certain Cr-Mo-Al and Cr-Mo-V steels, when properly heat treated, perform better in severe production service in sour-gas fields than does the conventional API Grade N-80 steel, and the same superiority is exhibited in laboratory sulfide-corrosion tests. (102,103,104) A minimum susceptibility to cracking in constructional Cr-Mo-V steels occurs after tempering the steel at elevated temperatures so as to produce a structure of fine spheroidal carbides. (94) Ferritic steels containing about 12 percent chromium are less susceptible to general corrosion than are steels with only 2 percent or so chromium, but they also show sulfide-corrosion cracking when heat treated to high strength levels. A 9 percent nickel steel was particularly susceptible to this type of cracking. (94) Fraser and co-workers(105, 106) studied 104 different alloys in the laboratory and under field conditions and derived empirical equations for rating the cracking susceptibility from composition, yield-strength, ductility, hardness, and heat-treatment data.

Water is an essential factor in the corrosion of iron by H_2S . (107) Dry H_2S and saturated solutions of H_2S in benzene do not result in cracking of stressed steel specimens. (101) Various investigators have observed that the corrosion rate of steel in dry H_2S at elevated temperatures is minimized by the presence of oxygen, unlike acid solutions where the presence of oxygen accelerates corrosion. The rate of sulfide corrosion may be reduced considerably by lowering the acidity of the environment, even though the same amount of sulfide is present; (108) however, this apparently occurs only in the presence of oxygen. (109,110) Cathodic protection and galvanic contact with less

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noble metals frequently are not adequate solutions to the problem of sulfide corrosion, because more hydrogen, not less, is supplied to the steel under these conditions. A little information is available to indicate that inhibitors can reduce the corrosion and lower the chance of failure of steels when stressed and immersed in H2S solutions. (111,112,93) It has been suggested that copper coatings might present certain advantages. (108) Coating with aluminum (either aluminizing or calorizing) generally is considered to protect steel sufficiently against attack by H₂S at high temperatures, but it is doubtful whether these coatings will prevent delayed cracking under the influence of moist H_2S . Warren and Beckman⁽⁹³⁾ found that a nickel electroplate 0.127 mm (0.005 inch) thick provided protection for heat-treated AISI 4140 steel bolts against sulfide-corrosion cracking, but coatings of cadmium, zinc, lead, and brass did not give reliable or consistent protection.

A number of other corrosion reactions can introduce hydrogen into steel. For example, Bloom(113) has described the corrosion-generated hydrogen cracking of martensitic and other hardened stainless steels; Hudson, Norris, and Stragand(81) found the fairly high hydrogen content of 7.6 cc/ 100 g of steel after treatment for 4 hours in a 1M solution of K2S2O8; and Uhlig(114) found that hardened 13 percent chromium steel failed rapidly when coupled to aluminum in 3 percent NaCl solution (a galvanic couple). Steigerwald(115) carried out an investigation to systematically evaluate the influence of liquid environments (primarily aqueous) on the delayed-failure characteristics of high-strength steel in the presence of very sharp notches (fatigue cracks). All the aqueous solutions and all the nonaqueous solutions except carbon tetrachloride (the only solvent that contained no hydrogen) consistently caused delayed failure. However, the times to failure for the nonaqueous solvents were considerably longer than those produced by the aqueous environments. The solvents with closed-ring structures and low dielectric constants, benzene and carbon tetrachloride, were capable of sustaining the applied load for the greatest periods of time.

It is possible that protective coatings such as zinc, cadmium, or aluminum may result in the introduction of hydrogen into steel during service. Since these metals are more anodic (that is, more active) than steel under most service conditions, at breaks in the coating the steel could become a cathode in an electrolytic cell, and this might result in hydrogen being introduced into the steel. This is one form of galvanic corrosion.

Entry by Electrochemical Processes

Cathodic protection is based on the principle that if an anodic current promotes corrosion, then a suitable cathodic current will prevent it. Gathodic protection of steel is accomplished in two ways. In the sacrificial-anode system, no external current is required. Anodes of zinc or of a magnesium or aluminum alloy are connected by a conductor to the steel structure to be protected. Electrochemical action alone (whereby the anode is slowly consumed) generates an electric current and provides the potential needed to protect the structure. With nonsacrificial anodes (such as a large ground bed of graphite for large steel structures), an external current must be supplied. With either system, atomic hydrogen is liberated at the surface of the steel (the cathode). Upon entering the steel, this atomic hydrogen is a potential source of hydrogen cracking or of blistering.

When steel is cleaned cathodically, machined electrolytically, electroplated, or purposely charged with hydrogen by cathodic treatment, cations of hydrogen are liberated on the steel surface (the cathode). Since free atoms or charged ions of hydrogen can exist at the steel surface only briefly, either they enter the metal lattice quickly or they form hydrogen molecules that cannot enter the steel. High hydrogen concentrations can be obtained in steel from these various processes, frequently with serious consequences. Furthermore, the presence of certain elements or compounds in the electrolyte, which are known as cathodic poisons, greatly increases the likelihood that hydrogen will enter the steel. Their presence is an aid to the laboratory worker who is studying the effects of hydrogen in steel and desires to charge the steel with hydrogen under conditions that are as reproducible as possible. However, in producing parts for service, it is necessary to take steps to minimize hydrogen entry in such processes as those listed above, and to remove most of that hydrogen which is picked up during processing.

The driving force for hydrogen entry when atomic hydrogen is liberated by electrochemical processes is the fugacity of hydrogen at the steel surface. This has been discussed previously in the section that deals with entry by corrosion processes.

Cathodic Charging. Cathodic charging is the method most widely used to charge specimens with hydrogen for the purpose of studying the delayed-brittle-failure phenomenon. Numerous examples of this procedure have been discussed in a previous report ty the present authors (DMIC Report No. 196). One of two procedures usually is followed. A notched specimen may be charged under standard conditions (standard electrolyte, cathodic poison, current density, time, and specimen geometry) and then placed under load when a fixed time has elapsed after cathodic charging. With smooth specimens, it usually is not possible to obtain failures when they are charged prior to being placed under load. This type of specimen may be studied by charging under standard conditions continuously while under load. In either case, the hydrogen content may be varied by changing the current density. With prior charging, the hydrogen content also can be varied by using different charging times and by holding for different times after charging but prior to placing the specimens under load, allowing hydrogen effusion to various degrees. Baking treatments also may be used to vary the hydrogen content. All of these procedures have been discussed in the previous report in regard to their effect on delayed brittle failures of steel.

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Some variables in the cathodic charging process, such as the effect of specimen thickness, will be discussed in the section of the present report on diffusion. Others, such as current density and cathodic poisons, are especially important to cathodic charging and are related to hydrogen entry, and, therefore, are considered here.

Possible Mechanisms. Chapter 4 of Smialowski's book(116) is devoted to a discussion of the cathodic evolution of hydrogen on metals which occlude this element. According to Smialowski, the electrolytic evolution of hydrogen from <u>acid</u> electrolytes consists of two main stages:

> Discharge of hydrated protons (the hydronium ion) and adsorption of neutral hydrogen atoms

$$H_{3}O^{+} + e \longrightarrow H_{ads} + H_{2}O.$$
 (15)

 (2) Desorption, that is, detachment of hydrogen atoms from the cathode and production of molecular hydrogen. Two alternative mechanisms are considered probable. Either the hydrogen atoms combine directly

$$H_{ads} + H_{ads} \longrightarrow H_2,$$
 (16)

or a hydrated proton discharges near an adsorbed hydrogen atom and gives molecular hydrogen as follows:

$$H_{ads} + H_3O^+ + e \longrightarrow H_2O + H_2.$$
 (17)

Reaction 16 is commonly called "catalytic desorption", because its rate is generally believed to depend upon the catalytic activity of the metal surface for recombination of hydrogen atoms into molecules. Reaction 17 is called "electrochemical desorption".

Either the discharge of hydrogen ions (15) or one of the two possible desorption mechanisms (16) or (17) determines the velocity of the overall process of hydrogen evolution from acids.

In <u>neutral and alkaline solutions</u>, water molecules supply the hydrogen evolved at the cathode. Smialowski writes the sequence of cathodic processes as follows:

 $H_2O + e \longrightarrow H_{ads} + OH^-$ (18)

$$H_{ads} + H_{ads} \longrightarrow H_2$$
 (19)

As in the preceding case, one of the above stages may be slow and rate determining.

Hydrogen evolved cathodically from alkaline electrolytes, such as NaOH, shows much less ability to enter iron and steel specimens than that evolved from acid electrolytes or neutral solutions.

Barton⁽¹¹⁷⁾ recently discussed in considerable detail the mechanisms whereby hydrogen finds its way across solution-metal interfaces and undergoes a change in its oxidation state. As a first approximation, Barton views hydrogen entry as a side reaction to the evolution reaction. Therefore, the various mechanisms of hydrogen evolution are considered in detail. Then the factors, such as surface coverage, which may affect the rate at which hydrogen enters the metal are considered. Finally, the factors involved in hydrogen release from metals are discussed.

It is unusual for a chemical reaction to occur in a single step. Generally, a number of consecutive or parallel steps or, frequently, a complex combination of parallel and consecutive steps constitute the mechanism. One of these steps will be the slowest, and it is called the rate-determining step. If it is enough slower than all the preceding or following steps so that the products and reactants of the other steps can attain equilibrium, it will completely control the rate of reaction.

In the hydrogen-evolution reaction, a number of steps are necessary in order to transfer or transform a proton from some proton-containing species in the bulk of the electrolyte solution to a hydrogen molecule in the gas phase. The following series of consecutive steps is a simplified presentation of the main features of the process.

- A proton-containing species must diffuse from the bulk of the solution to the outer boundary of the double layer surrounding the electrode. Hydrogen discharge, particularly from aqueous solutions, usually is not diffusion controlled at this step.
- (2) The proton must be transferred across the charged double layer; this is probably a diffusion-controlled process.
- (3) The proton species then may be adsorbed on the electrode. The deleterious effects of impurities on the hydrogen-electrode reaction, which frequently plague research in this area, often result from specific adsorption of impurities on the metal surface that results in a blocking action to either the adsorption of the reactant or to a subsequent step. However, an adsorbed layer does not necessarily prevent hydrogen reduction.

- (4) The charge transfer from the electron sea of the metal to the proton may be the slow step in the process in some cases. This reduction step leaves an adsorbed layer of hydrogen atoms on the electrode surface. These atoms either enter the metal or combine to form hydrogen molecules which are not capable of entering the metal.
- (5) The direct combination of two adsorbed atoms to form an adsorbed hydrogen molecule has been the traditionally accepted slow step in the evolution reaction. Another combination process, the electrochemical one, involving the simultaneous combination of an adsorbed atom, a proton, and an electron from the metal has been shown to be important in many cases.
- (6) The hydrogen molecules must be desorbed and diffuse away from the electrode surface. If this is a slow step, the hydrogen molecules can have a blocking effect on the active electrode area, with a consequent reduction in the rate of either the adsorption or the charge transfer.
- (7) The final step, at least under conditions of high current density, is the formation of bubbles.

Many mechanisms have been proposed to account for the charge transfer, atom formation, and molecule formation. Barton discusses the few that have been widely accepted, both for acid and alkaline electrolytes. The equations for these reactions are listed in Table 7.

In <u>highly acid solutions</u>, where the source of of the proton is H_3O^+ , these three mechanisms have gained wide acceptance:

- Rate determination by atomic combination.
- (2) Rate determination by proton discharge. This consists of a slow discharge step followed by one of two fast combination steps (2a and 2b).
- (3) Rate determination by electrochemical desorption. For the slow electrochemical desorption, there are two cases, one with low surface coverage and one with complete coverage.

In strongly alkaline solutions, the same three mechanisms also occur, but with the proton source being H_2O , along with one additional mechanism involving the solution cation. Step 4 in alkaline solutions is rate determination by alkali-metal amalgam reaction with water, followed by either

$$H_{ads} \xrightarrow{fast} H_2$$
 (21)

Acid Solutions

- (1) $H_3O^+ + e \longrightarrow H_{ads} + H_2O$ $H_{ads} \xrightarrow{s} H_2$
- (2a) $H_3O^+ + e \xrightarrow{s} H_{ads} + H_2O$ $H_{ads} \xrightarrow{} H_2$
- (2b) $H_3O^+ + e \xrightarrow{s} H_{ads} + H_2O$ $H_{ads} + H_3O^+ + e \xrightarrow{g} H_2 + H_2O$
- (3) $H_3O^+ + e \xrightarrow{H_{ads} + H_2O} H_{ads} + H_3O^+ + e \xrightarrow{E} H_2 + H_2O$

Alkaline Solutions

- (1) $H_2O + e \longrightarrow H_{ads} + OH^ H_{ads} \xrightarrow{s} H_2$
- (2a) $H_2O + e \xrightarrow{s} H_{ads} + OH^ H_{ads} \xrightarrow{} H_2$
- (2b) $H_2O + e \xrightarrow{s} H_{ads} + OH^ H_{ads} + H_2O + e \xrightarrow{} H_2 + OH^-$
- (3) $H_2O + e \longrightarrow H_{ads} + OH^ H_{ads} + H_2O + e \longrightarrow H_2 + OH^-$
- (4) $M^{+Z} + Ze \xrightarrow{M} M$ M + ZH₂O $\xrightarrow{s} M^{+Z} + ZH_{ads} + ZOH^{-1}$

or

$$H_{ads} + H_2O + e = \frac{fast}{H_2} + OH^-$$
. (22)

The mechanisms of hydrogen evolution in both strongly acid and strongly alkaline solutions are fairly well known for many metals. Knowledge is much less complete for the region of intermediate pH. The reader is referred to Barton's paper for a discussion of each of these mechanisms and for additional references.

It has been realized for some time that the nature of the cathode surface is of prime importance in determining the mechanism of the hydrogen-electrode reaction and the mode of hydrogen entry. The nature of the surface is not well understood, even though some work has been done in which the surfaces have been prepared carefully enough to give reproducible results. The rate of the hydrogen-electrode reaction depends upon such variables as alloy composition, porosity, heat treatment (including annealing), and surface treatments such as electropolishing or anodic activation. Appreciable differences occur between different crystallographic faces prepared in the same way.

Barton analyzed the entry process as a side reaction to the evolution process, assuming that one of the hydrogen-containing species present in the evolution process either is or leads to the entry species. It is to be expected that the entry species would be adsorbed on the surface and that it is present in reasonably high concentrations. The mechanisms listed previously are associated with various amounts of hydrogen atoms adsorbed on the surface. Barton considered each one in turn. If it is hypothesized that the entering species is adsorbed hydrogen atoms, the entry rate should correlate with the extent of surface coverage by hydrogen. Data are presented which support this hypothesis. Also, the low coverage in alkaline solutions compared to acid solutions is reflected in the relative permeabilities of hydrogen introduced into steel from these solutions, as will be discussed in the section dealing with the electrolyte. The concentration of hydrogen molecules adsorbed on the surface is believed to be negligible for all mechanisms for iron and steel. Thus, molecular hydrogen does not appear to be a likely species to supply the entering hydrogen. This is supported by gas-phase studies (referred to previously) which indicate that the species in the metal is monatomic. Also, Barton presented arguments against the hypothesis that adsorbed molecular hydrogen and water furnish the entering hydrogen.

If an approximately linear relation exists between the overvoltage and the fraction of the surface that is covered with hydrogen, and if the Tafel relation between overvoltage and current density is employed, the coverage should vary as the log of the current density and so should the permeation rate if coverage is the determining factor. As will be discussed in the next section, the dependence of permeation rate on current density has been investigated many times and usually is given as a square-root function.

It has been suggested that the effects of mechanical processes such as shot peening, polishing, or grit blasting possibly should not be attributed to mechanical working of the metal surface layer but may be due to alterations in the state of the surface.

Current Density. One of the major variables in the entry and permeation of cathodically evolved hydrogen in and through steel is the current density. In 1922, Bodenstein found that permeation of hydrogen followed the relationship

$$P = k i^{\frac{1}{2}},$$
 (23)

where i is the cathodic current density and k is a constant. The relationship between permeation and current density is the same as the relationship between permeation and pressure given earlier in the present report. Because of this similarity, a number of investigators concluded that the phenomena of hydrogen flow through metals from the gas phase (pressure charging and thermal charging) and from an electrolyte (electrolytic charging) are governed by identical laws and have similar mechanisms. However, a permeation rate that is attained readily at rather low values of the cathodic current density may be attained by pressure charging only with very large pressure differences, frequently thousands of atmospheres.

A number of investigators have observed hydrogen behavior different from that expressed by Equation 23. For example, Elsea and coworkers(118) performed a series of experiments with high-strength SAE 4340 steel in which the composition of the electrolyte, the current density, and the thickness of the permeation specimen were varied. Part of the results of these experiments are shown in Figure 14. In this figure, the log of the steady-state permeation rate is plotted as a function of the log of the current density for two different electrolytes, 4 percent H2SO4 with phosphorus poison and 1/2 percent NaOH containing no cathodic poison. In all cases, the data fit a straight line reasonably well. Thus, an equation of the form

$$P = ki^n$$
 (24)

is indicated. Since the experiments were performed under steady-state permeation conditions, in which the quantities of hydrogen entering and leaving the specimen would be the same, this equation also expresses the relationship between entry rate and current density. The variation in the slopes of the lines shows that the exponent n was not constant. It varied from about 0.54 for the 0.040-in. -thick specimen charged in the H_2SO_4 electrolyte to 1.0 for the specimen of the same thickness that was charged in the NaOH electrolyte and a little over 1.0 for the thinner specimen charged in NaOH.

At very low current densities, Raczynski (as described by Smialowski(119)) found that the permeation rate of hydrogen through an iron membrane was proportional to the current density, whereas at somewhat higher current densities it was proportional to the square root of the current density. In his plot of permeation rate versus the square root of the current density (see Figure 15), he identified four regions as follows:

> Region a, in which P is nearly proportional to i

Region b, in which P is proportional to $i^{\frac{1}{2}}$

Region c, in which a more complicated relationship exists between P and i

Region d, in which P is independent of i.



FIGURE 14. PERMEATION RATE OF HYDROGEN AS A FUNCTION OF CURRENT DENSITY FOR BOTH 4 PERCENT H₂SO₄ AND ¹/₂ PERCENT NaOH ELECTROLYTES AND FOR VARI-OUS THICKNESSES OF SPECI-MENS(118)

(SAE 4340 steel.)

Permeation Rate

FIGURE 15. PERMEATION RATE VERSUS SQUARE ROOT OF CURRENT DENSITY (AFTER RACZYNSKI)⁽¹¹⁹⁾

Raczynski showed that the d-region for mild steel may be reached at different values of the minimum current density. The critical currentdensity value, at which the permeation rate begins to be independent of the current, density, is lower for thicker membranes, at lower temperatures, and in the presence of cathodic poisons. For annealed mild-steel membranes 0.1 and 0.2 mm thick, charged cathodically at 68 F (20 C) in 1N H₂SO₄ containing As₂O₃ cathodic poison, the following critical values were obtained:

Thickness,	Critical Current		
mm	Density, amp/cm ²		
0.1	0.15		
0 2	0.05		

The constant permeation rates for higher current densities are shown in Figure 16. For a given membrane, the effect of temperature on the critical value of the current density for attaining constant permeation rate was as follows:

Temperature		Critical Current
С	F	Density, amp/cm^2
30	86	0.15
80	176	0.35

Although Borelius and Lindblom⁽⁷²⁾ obtained data which suggested that there is a threshold value of current density below which no permeation

Since this analysis was for steady-state permeation, the hydrogen-entry rate showed the same relationships with current density. Barrer⁽¹²⁰⁾ also showed that at high current densities it is possible to reach a condition such that the rate of entry (permeation rate under steady-state conditions) no longer increases as the current density is increased. According to Barrer, it is necessary to achieve Raczynski's Condition d if a diffusion coefficient is to be determined from permeation data.



FIGURE 16. PERMEATION RATE VERSUS CUR-RENT DENSITY FOR 0.1- AND 0.2-MM-THICK MEMBRANES (AFTER RACZYNSKI)⁽¹¹⁹⁾

is possible, the majority of later investigators failed to confirm its existence. For example, according to Smialowski, ⁽¹¹⁹⁾ Raczynski observed permeation through an iron membrane 0.1 mm thick at a current density of only 0.2 ma/cm².

For a hardened alloy steel cathodically charged in a very weak acid electrolyte (buffered solution of pH = 4.0) at low current densities, Bolton and Shreir⁽¹²¹⁾ found a linear relationship between the hydrogen content (not the entry rate) and the logarithm of the current density, at least for current densities between 0.1 and 2.0 ma/cm². These results, which were obtained for a fixed charging time which in all cases was less than that required to reach saturation, are shown in Figure 17.



FIGURE 17. RELATIONSHIP BETWEEN HYDROGEN ABSORBED AND LOG i (HARDENED REX 539, ACETATE BUFFER pH 4.0, POLARIZED 3 HR)(121)

Electrolyte. Apparently, quantitative information regarding the ability of hydrogen cathodically evolved from different acids to penetrate into iron are not available. Smialowski(119) suggests that the following is probably the decreasing order for the strong inorganic acids: orthophosphoric, sulfuric, and hydrochloric. Nitric acid, an oxidizing acid, has long been regarded as causing no hydrogenation. However, as was discussed in the section on acid pickling, Hudson, Norris, and Stragand⁽⁸¹⁾ found that nitric acid can result in hydrogen being charged into steel, at least under some conditions.

Many investigations have been carried out on absorption of hydrogen from strongly acid and strongly alkaline solutions. Many of these investigations have been for the purpose of developing cleaning and electroplating procedures to minimize hydrogen absorption during electrolysis. Also, a number of workers have purposely introduced hydrogen into steel by cathodic charging in strong acid solutions so as to determine the effects of hydrogen on mechanical properties and on the propensity toward delayed brittle failure under static load. Many of these investigations have been described in a previous report on hydrogen in steel by the present authors (DMIC Report No. 196). When the acid contains a cathodic poison, large amounts of hydrogen (far above the solubility at room temperature) can readily be charged into steel. The importance of such factors as the nature of the surface of the cathode and the presence of impurities in the electrolyte often have been overlooked in such studies, with the result that reproducibility has not always been what might be desired. The presence of the poison in the electrolyte is of such importance that it will be considered separately in the next section of this report. The problems of accurately determining the quantities of hydrogen absorbed are so great that most investigators have evaluated the results of charging variables by their effect on mechanical properties, and particularly, on the susceptibility to delayed brittle failure.

In most of these experiments, it was desired that hydrogen be introduced into the steel at an appreciable rate. Therefore, the composition of the acid electrolyte was not of too much importance so long as it was constant for a given series of experiments. The liberation of hydrogen at the surface of the specimen was controlled by controlling the current density, as was discussed above. However, in experiments in which it is desired to introduce hydrogen at low rates so as to simulate conditions that could arise in service (for example, cathodically protected structures or steel coated with galvanically less noble metals), the composition of the electrolyte is highly important. At the very low current densities necessary, it is difficult to find an electrolyte that will provide cathodic protection for the specimen (the cathode). If cathodic protection is not maintained, corrosion

of the specimen will introduce an additional and uncontrolled amount of hydrogen. One approach is to use weakly ionized acids, such as acetic acid, oxalic acid, or lactic acid. To provide sufficient conductivity, it may be necessary to substitute a solvent such as ethylene glycol for the water used conventionally. One problem that the present authors have found in their investigations with such electrolytes is that side reactions may occur at one or both electrodes. Frequently, therefore. the amount of hydrogen liberated at the specimen surface is far less than is calculated from the time and current, using Faraday's law. The authors have successfully used an electrolyte consisting of three parts by volume ethylene glycol and one part of a 5 percent (by weight) aqueous oxalic acid solution with mild steel at a current density of 1 ma/ in.² in a number of experiments.

Hydrogen also can be introduced into steel by electrolysis in alkaline electrolytes. Such electrolytes offer the advantage of providing cathodic protection at low current densities. However, low permeation rates are obtained when they are used. (118) The relationship between hydrogen absorption and cathode overvoltage is different for acid electrolytes and alkaline electrolytes. This behavior probably is the result of a change in the mechanism of the hydrogen-evolution reaction in this range where the pH at the cathode may be sufficiently high for water as well as the hydronium ion, $(H_3O)^+$, to be reduced to hydrogen. (121) Barton(122) suggested that the tendency for some metals to change to a slow-hydrogen-discharge mechanism in alkaline solutions with a resultant low coverage of the specimen surface may also explain the lower entry rate of hydrogen at high pH values.

The common effects observed at a cathode during electrolysis are (1) the discharge of a positive ion when the ion is below hydrogen in the electromotive series, (2) the formation of a negative ion from a reducible material such as chlorine gas, (3) the reduction of an insoluble salt with the formation of a negative ion into solution, and (4) the evolution of hydrogen gas when the positive ion lies above hydrogen in the electromotive series. In <u>acid solutions</u>, electrolysis is accompanied by the loss of hydrogen ions from solution as shown by the reaction

$$H^{+} + e^{-} = H^{\circ},$$
 (25)

and in <u>neutral or alkaline solutions</u>, it is accompanied by the formation of hydroxyl ions and the decomposition of water as shown by the reaction

$$H_2O + e^- = H^\circ + OH^-$$
. (26)

Thus, one would expect the entry and permeation of hydrogen to be different when electrolysis is carried out in acid solutions than it is when carried out in neutral or alkaline solutions. The mechanisms of cathodic evolution of hydrogen on metals has been discussed in a previous section of the report. Poisons and Inhibitors. Some elements of Groups V andVI of the periodic table act as poisons or "negative catalysts" for the hydrogen recombination reaction. These are phosphorus, arsenic, antimony, sulfur, selenium, and tellurium. There are indications that the elements bismuth, mercury, lead, and tin and the CN⁻ ion have a similar effect. By slowing down the rate at which hydrogen atoms combine to form molecular hydrogen and thus increasing the lifetime of the atoms, these catalytic poisons promote the entry of hydrogen into iron, steel, and other metals during cathodic charging or pickling. Alexeev and Polukarov reported this effect in 1926 and apparently were the first to observe it.

Smialowski and Szklarska-Smialowska(123,124) established the following order for decreasing ability to promote hydrogen entry into iron cathodes: phosphorus, sulfur, arsenic, selenium, antimony, tellurium, and bismuth. The anions SO₄⁻² and PO₄⁻³ have a high degree of oxidation and stability and are not reduced at the cathode. Therefore, they do not poison the hydrogen recombination process at the steel surface. However, lower oxides, such as SO₂, and oxides and oxidized anions of other elements mentioned above are reduced at the cathode to give free elements and hydrides. For example:

According to Smialowski, (119) the formation of a volatile hydride is essential for the poisoning effect. Also, certain organic compounds, for example thiourea, decompose at the cathode and form H₂S, a potent catalytic poison. It has been demonstrated that part of the lack of reproducibility of results of various investigators using cathodic charging has been due to the fact that sulfuric and hydrochloric acids, unless they have been especially purified, usually contain traces of arsenic, selenium, and other elements that act as cathodic poisons. Smialowski and Szklarska-Smialowska(125) found that arsenic distinctly promotes the penetration of hydrogen into iron cathodes when present at a concentration as small as 4×10^{-8} mole/1. Selenium is effective when as little as 3×10^{-6} mole/l is present. Since during prolonged electrolysis these elements escape into the atmosphere as hydrides, the cathodic poison content of the solution may be depleted.

Specimens of iron and the various steels always contain a certain amount of phosphorus and sulfur. If the specimen is allowed to contact the acid before the current is turned on, the acid is invariably contaminated. The extent of contamination depends on such factors as the composition and structure of the sample, the acidity of the solution, the elapsed time before the external current is applied so that the sample becomes cathodically protected, and the temperature.

The rate of hydrogen entry is very low when iron or steel is charged cathodically in pure acid

solutions; however, when a cathodic poison is added to the acid electrolyte, the quantity of hydrogen absorbed is much greater. (28,123-128) This is illustrated by some of the results Smialowski and Szklarska-Smialowska(28,124,125) obtained in their study of the effects of cathodic poisons, performed in conjunction with studies of the dimensional changes in iron during cathodic charging in acids. Figure 18 summarizes the results obtained in one experiment in which the elongation of annealed mild-steel wires was measured as a function of the time of cathodic charging. The first portion of Curve 1 shows the behavior of the steel wire when treated cathodically for 310 minutes in pure 1N H2SO4 solution at a current density of 0.15 amp/ cm^2 and at a temperature of 37 F (3 C). The rate of elongation resulting from hydrogen penetration was comparatively low under these conditions. However, the rate rose markedly after an addition of 1 g of thiourea per liter of solution was made in the 311th minute (indicated on the curve as Th. u.). Thiourea is decomposed at the cathode and H2S is formed which acts as a poison and results in increased penetration of hydrogen into the steel cathode, the penetration being accompanied by plastic deformation of the wire. Curve 2 represents the behavior exhibited by a wire treated for 130 minutes in IN H₂SO₄ that contained 0.16 g dibenzyl sulphoxide per liter. This compound is a corrosion inhibitor which becomes strongly chemisorbed on the surface of the polarized cathode at certain potential values, and the penetration of hydrogen into the iron cathode is greatly reduced. The subsequent addition of thiourea caused only a slight increase in hydrogen penetration rate when dibenzyl sulphoxide was present in the electrolyte. Curve 3 shows the behavior in acid containing 1 g As₂O₃ per liter at 86 F (30 C). Curve 4 shows the elongation of the wire in the same arseniccontaining electrolyte at 122 F (50 C), but 5 minutes after the beginning of the electrolysis, 0.1 g dibenzyl sulphoxide (dbs) per liter of solution was added. This caused an immediate drop in the elongation rate. Different organic compounds produce effects like one or the other of these two compounds, thiourea and dibenzyl sulphoxide. Therefore, the electrolyte should always be maintained free from organic impurities if reliable experimental results are to be obtained.

Because the elongations in a straight wire were difficult to measure, these investigators developed another technique of coiling up a long length of wire for immersion in the electrolyte. Elongation of the coil (not the wire itself) was much larger and easy to measure. Using this technique they studied the effects of the different poisons on the deformation effect associated with hydrogen absorption. The experiments were performed with mild-steel wire, annealed after winding. The coils did not exhibit any deformation when subjected to cathodic conditions in a solution of pure H_2SO_4 . However, when the electrolyte contained a cathodic poison, rapid elongation of the coil occurred until a limiting elongation was reached.



FIGURE 18. ELONGATION OF IRON WIRE AS A FUNCTION OF TIME OF CATHODIC CHARGING⁽¹²⁵⁾

(Original length of wire = 180 mm.)

Sometimes the elongation of the <u>coil</u> amounted to as much as 10 times the original coil length (actually a very small elongation of the wire itself). The effect of poison concentration is illustrated by the results obtained for arsenic, shown in Figure 19. Other data they obtained showed that the poison content of the electrolyte must exceed some rather critical amount, if the poison is to be very effective.

Figure 20 shows the relationship these investigators found between the limiting values of elongation and the atomic number of the elements from Groups V-B and VI-B of the periodic system used as poisons. These data were obtained from coils with a cathodic current density of 0.036 amp/ cm^2 at 68 F (20 C). The concentration of additions in the 1N H₂SO₄ electrolyte was always greater than the critical amount. Bismuth, selenium, tellurium, and antimony were added in the form of oxides, sulfur was added as H₂S, and phosphorus was added in the form of hydrogen phosphide. Neither SO₄⁻² or PO₄⁻³ ions caused penetration of hydrogen into iron cathodes, as measured by the elongation of coils.

It appears that a relationship exists between the deformation effect observed in the presence of any given poisoning element in the electrolyte and the binding force of the respective hydride that was computed by Siebert (see Table 8), (125) Oxygen and nitrogen, which form compounds with binding-force constants higher than in the hydrogen molecule itself, exhibit no measurable poisoning effect. Among the other elements listed, the strongest bond is with the sulfide ion, and it resulted in the greatest deformation as the result of hydrogen penetration of the steel cathode. Antimony makes a relatively weak bond and had a feeble poisoning effect. Selenium showed a weaker poisoning effect than was expected from its bindingforce constant.



- FIGURE 19. ELONGATION OF IRON COILS CHARGED CATHODICALLY IN 1N H₂SO₄ WITH THE ADDITION OF DIFFERENT AMOUNTS OF ARSENIC EXPRESSED AS mg As/1 BUT ADDED AS As₂O₃⁽¹²⁵⁾
 - (Original length of coil immersed in acid = l_{\circ} = 50 mm. Length of wire in coil = L = 1000 mm.)

Darken and Smith⁽⁴⁸⁾ also studied the effect of inhibitors on hydrogen permeability and found a marked effect. Only two inhibitors--quinoline ethiodide (0.1 percent) and Nep No. 22--were investigated, both used with $1N H_2SO_4$. Both of these inhibitors reduced the permeability of several steels to very low values, compared to those obtained with the uninhibited acid.

Structure. From measurements of overvoltage, Bolton and Shreir⁽¹²¹⁾ obtained data which suggested that it is easier kinetically to discharge hydrogen on hardened steel tempered at low temperatures than on the same material tempered at high temperatures (see Figure 21). This, they stated, was in agreement with the results of galvanostatic (constant current) experiments, the results of which are shown in Figure 17. However, for a constant current density, the specimens absorbed hydrogen in the order of increasing tempering temperature, that is, martensite-like structures had a lower hydrogen absorption rate than did tempered structures. This is shown in Figures 17 and 22. For each of the tempered structures investigated, the relationships between the amount of



FIGURE 20. LIMITING ELONGATION OF IRON COILS VERSUS ATOMIC NUMBER OF THE ELEMENT USED AS POISON DUR-ING CATHODIC POLARIZATION(125)

TABLE 8. COMPARISON OF THE LIMITING ELONGATION OF IRON COILS WITH THE BINDING FORCE OF THE HYDRIDES(125)

Element	Atomic Number	Valence (n)	Hydride	Binding Force Constant ,dynes/Å	Mear. Limiting Elongation, mm (41)	Elongation per Atom H, $\frac{\Delta \lambda}{n}$
н	1	1	H ₂	5.76		
N	7	3	NH3	7.00		
O P S	8 15	2	H ₂ O	8.41		
P	15	3	PH3	3.47	148	49.3
S	16	2	H ₂ S	4.31	121	60.5
As	33	3	AsH3	2.91	111	37.0
Se	34	2	H ₂ Se	3.31	68	34.0
Sb	51	3	SbH3	2,26	84	28.0
Te	52	2	HzTe	2	42	21.0
Bi	83	3	BiH ₃	2	2	0.7



FIGURE 21. OVERVOLTAGE-TIME CURVES AT CONSTANT CURRENT⁽¹²¹⁾



hydrogen absorbed and the cathode overvoltage and, likewise, hydrogen absorption and the log of the current density were linear over the limited range of conditions investigated.



TURE ON ABSORPTION OF HYDRO -GEN BY HARDENED STEEL⁽¹²¹⁾

Electroplating. The ductility of steel may be markedly reduced and its propensity toward delayed brittle failure under static load may be markedly increased as the result of an electroplating process. These effects result from hydrogen that is plated concurrently with the metal, part of the hydrogen being absorbed by the steel. This has been a serious problem with high-strength steel parts. The electroplated coating is required for corrosion resistance, yet its application makes the part susceptible to failure by another mechanism - hydrogen cracking. Furthermore, the presence of the plate makes the removal of hydrogen from the steel much more difficult.

Early studies showed that the embrittlement becomes more severe as the hardness level of the steel is increased. (129,130,131) Steel is embrittled by virtually all of the electroplating processes, including cadmium, chromium, zinc, tin, nickel, lead, copper, and silver. (132-139) Hydrogen introduced into specimens during cadmium electroplating may be as high as that introduced through severe cathodic charging. (139) One investigation showed that the delayed-failure behavior was almost identical for commercial cadmium-plated steel and the same steel cathodically charged with hydrogen under a standard charging condition that was much used in studies of hydrogen-induced delayed brittle failure. (140) As a result of these various investigations, it is apparent that sufficient hydrogen is introduced into steel during a commercial-type electroplating process to cause substantial embrittlement and to cause delayed brittle failure of steel under many conditions. Other work has shown that frequently more hydrogen is introduced in acid pickling to remove scale or in cathodic cleaning prior to electroplating than is introduced in electroplating itself. ⁽¹⁴¹⁾

The method for estimating the hydrogen input to a workpiece during electrolysis that has been used most widely consists of determining the "efficiency" of the plating bath. This is done by measuring the quantity of electricity passed during plating, then using Faraday's laws of electrolysis to determine what weight of metal should have been plated on the cathode. By comparing the weight of metal actually deposited with that predicted by Faraday's laws, a measure of the efficiency of the plating bath being monitored is obtained. Efficiencies of 99.8 percent have been obtained. If it is assumed that the remaining 0.2 percent of the plating current deposits hydrogen instead of metal, it has been calculated that the amount of hydrogen plated at a plating current density of 50 amperes per square foot is approximately 6 x 1017 hydrogen atoms per square foot per second. This method suffers from being low in its sensitivity to hydrogen estimation, and other drawbacks have been pointed out by McGraw et al(142) as a result of their study of hydrogen entry into metals during chemical and electrochemical processing.

Some investigators formerly attempted to evaluate hydrogen absorption during plating or the hydrogen remaining after a baking treatment indirectly by its effect on the tensile properties, particularly reduction in area. However, a number of instances have been reported in which hydrogen contents low enough to have no effect on the tensile properties have resulted in delayed brittle failure under static loads. After considerable study, it is now generally agreed that the sustained-load test of notched specimens is the only satisfactory mechanical test for evaluating hydrogen absorption. The common method of applying the test is to subject plated notched specimens to stresses of 70 to 90 percent of the notched tensile strength of unplated specimens for 100 hours or more. Rupture within 100 hours is generally taken as being indicative of embrittlement problems. However, brittle failures have occurred in triaxially stressed, cold-worked, plated test samples after 13,000 hours. (87) Thus, even this test is not sensitive enough or fast enough to be completely satisfactory for evaluating the absence of excess hydrogen in processed parts. These aspects have been discussed more fully in the section on test methods in a previous report by the present authors (DMIC Report No. 196).

Some very precise information on the entry and permeation of hydrogen through steels in contact with liquid media has been obtained using mass

⁽REX 539 steel, acetate buffer pH 4.0, overvoltage 100 mv, charging time 16 hr.)

spectrometers to count the ions of hydrogen as they are desorbed by the surface of the steel not in contact with the liquid. (91,46) However, the elaborate apparatus and the highly trained scientists required prohibit using this method to monitor plating lines. Use of the metal vacuum tube as a means of monitoring hydrogen, particularly when the recently developed active system(87) is employed, offers considerable promise for monitoring plating lines. The technique has been discussed a little more fully in a previous section of this report.

Lawrence(87) used the metal vacuum tube in an active system as a means of monitoring hydrogen absorption during the entire plating procedure. The results obtained with this hydrogen-detection gage were used to alter materially the lowembrittlement cadmium-plating process for highstrength steel. It was found that the rinses following pickling or plating added considerable extra hydrogen to the plated parts. The new process now requires that the parts be rinsed in overflow tanks, the rinse water being changed at least once each hour. A reduction of more than a factor of a thousand in hydrogen input rate during the plating process itself was achieved by lowering the voltage across the plating tank to 1 volt maximum. This was done with no decrease in cathode current density by raising the anode-to-cathode surface-area ratio to 6:1, and by keeping the anode-to-cathode distance small. By keeping sulfide concentration in the bath to a minimum of less than 1 ppm, both hydrogen sorption rates and sulfide stress corrosion were reduced greatly (sulfide is believed to be the most potent cathodic poison of the hydrogen recombination reaction). Deionized water with 0.5 ppm maximum impurity content is required both for rinses and for the plating solution. The porosity of the plating is judged both by observing the hydrogen sorption rate during plating and by leaving the plated tube in the plating solution after the plating current has been turned off. With porous plating, hydrogen input does not cease with current shut off. Because of the cadmium-iron galvanic cell, the presence of the electrolyte provides the proper coupling so that the cadmium dissolves and is replaced by hydrogen from the solution. With nonporous plating, the hydrogen input rate is so low that it is difficult to detect even with this sensitive hydrogen-detection gage.

It has been widely recognized for a decade now that hydrogen-induced delayed brittle failure, as a potential source of fallure of steel, is a serious problem. Failures most frequently occur under sustained-load conditions at locations where stress concentrations exist, and they may occur at stresses far below the yield strength of the material. Electroplating processes are recognized as being the most common source of this detrimental hydrogen, and cadmium electroplating processes are considered to be among the worst offenders. Hydrogen cracking became a serious problem when steels were first used in aircraft at tensile strength levels above about 200,000 psi, for structural parts such as landing gear and fasteners. Steels are used now at strengths up to at least 280,000 psi, and further increases in strength are anticipated. A few years ago, the Air Force found it necessary to prohibit cadmium electroplating of certain steels having strengths greater than 220,000 psi. Thus, it was necessary for manufacturers to resort to other methods of providing corrosion protection that were inferior to cadmium electroplating in this respect. Therefore, many studies of cadmium electroplating processes were carried out to determine if one or more could be used safely for electroplating ultrahigh-strength steel. The reader is directed to References (143), (144), (145), (146), and (147) for descriptions of some of these studies.

Among the cadmium electroplating baths studied were the conventional cadmium cyanide plating bath with and without brightener, the highefficiency cyanide plating bath with and without brightener, the cadmium fluoborate plating bath with various grain refiners and with none at all, and the cadmium sulfamate plating bath containing various grain refiners. Reference is made to the work of Geyer, Lawless, and Cohen^(145,146) to illustrate some of the results obtained. They used notched specimens of SAE 4340 steel heat treated so as to have an average notched tensile strength of 435,000 psi. All unplated specimens withstood 300,000 psi for 400 hours without failure, and the single specimen loaded at 400,000 psi withstood 100 hours without failure. A maximum applied stress of 300,000 psi was arbitrarily selected as the basis for a test used to determine if any detrimental hydrogen embrittlement had occurred in the plated specimens. A criterion was established that if one specimen of a group failed, then the entire group would be considered as having failed. All plated specimens were baked at 375 F, generally for 23 hours, prior to testing.

Detrimental hydrogen embrittlement was obtained from the conventional cyanide bath, as was expected. Most specimens failed to sustain even 220,000 psi. At least one specimen from each group plated in the fluoborate bath failed. No detrimental hydrogen embrittlement occurred when the high-efficiency cyanide bath was used without a brightener. Since the coating was granular in appearance, a brightener was added to the bath to improve the surface finish. It was shown previously that this bath with a brightener had a higher cathode current efficiency than the bath without a brightener. However, specimens plated with the brightener failed in the sustained-load test, so efficiency was not the controlling factor. It appeared that the significant difference in these two deposits with respect to hydrogen embrittlement was in the physical structure of the cadmium electrodeposits. Granular deposits have greater porosity than smooth deposits, thereby allowing the hydrogen to be driven out to a greater extent during the baking treatment after plating. No failures occurred in

1150 hours at 300,000 psi for specimens plated in the cadmium sulfamate bath with standard catechol grain refiner. However, the deposit was coarse grained and it was necessary to add additional grain refiners. When this was done, all specimens failed in 73 hours or less, even though the bath operated at approximately 100 percent cathode current efficiency over the entire plating range of current density, either with or without the additional grain refiners. Therefore, the pronounced differences in hydrogen embrittlement obtained with this bath were attributed solely to differences in the structures of the electrodeposits. Thus, it was shown that residual detrimental hydrogen embrittlement obtained from a cadmium electroplating bath is a function of cathode efficiency, which controls the entry of hydrogen, and of the structure of the coating, which controls the rate of removal of hydrogen during the postplating baking treatment.

A specimen coated with cadmium from a proprietary cadmium immersion solution (nonelectrolytic) failed after only 46 hours when stressed at 300,000 psi.

Four specimens were coated commercially with cadmium by a vacuum-metallizing technique. All withstood 300,000 psi for 300 hours without failure. Then three of them were exposed to salt fog for 192 hours, which resulted in light rusting. They again sustained a load of 300,000 psi for 300 hours without failure. The sustained-load test after exposure to salt fog was performed to determine if any detrimental hydrogen embrittlement resulted from the corrosion, since cadmium, being anodic to steel, would result in release of hydrogen at the cathodic areas during the corrosion process.

To determine whether cadmium is a barrier to hydrogen penetration, Geyer, Lawless, and Cohen took one of the specimens that had been vacuum metallized with cadmium and had been loaded for 300 hours at 300,000 psi without failure and subsequently electroplated it with cadmium from the fluoborate bath. The specimen was not baked after plating and broke upon loading to 300,000 psi. It was concluded that the vacuum-metallized cadmium did not serve as a barrier to hydrogen.

Apparently, cadmium can serve as a barrier to hydrogen in some instances. Perhaps this depends on the structure of the cadmium coating. Elsea and co-workers⁽¹⁴⁸⁾ performed an experiment to study the effect of cadmium plate on the absorption of hydrogen by steel during cathodic charging. A tubular specimen was cadmium plated on the exterior surface only and then was baked at 340 F to remove the hydrogen introduced by plating. Cathodic charging of this specimen for 24 hours with the poisoned sulfuric acid electrolyte in contact only with the cadmium-plated surface resulted in no measurable accumulation of hydrogen within the specimen. In another experiment, (118) a cadmium plate approximately 0,0003 inch thick was electrodeposited on the entrance side of permeability specimens. Both the cyanide and the fluoborate plating processes were used, and several specimens of each type of plate were tested. Without exception, when the cadmium plate was on the entrance surface or electrolytic cell side, there was no absorption of hydrogen, not even after several hours of charging with current densities of 10 to several hundred ma/in.² with the poisoned 4 percent sulfuric acid electrolyte, or 500 ma/in.² with the sodium hydroxide electrolyte. The behavior was the same whether the cadmium plate was applied by the cyanide or the fluoborate process. Thus, hydrogen did not diffuse through a cadmium plate at temperatures close to room temperature. Other investigators have shown that the dense cadmium plate that has been used widely for corrosion protection of high-strength steel parts markedly hinders effusion of hydrogen, and low hydrogen contents are not obtained by aging at room temperature or at the elevated temperatures normally used for baking treatments (about 375 F). However, a suitable porous cadmium plate permits hydrogen effusion while still providing appreciable corrosion protection.

Somewhat along the same line, investigators at Case Institute of Technology developed a cadmium electroplating procedure that consisted of plating a flash of cadmium on the specimen, baking to remove the hydrogen (which they claimed was feasible through such a thin cadmium layer), then plating to build the cadmium plate up to the desired thickness. This method did not provide complete insurance against failures and apparently has not been adopted commercially.

Beck and Jankowsky(147) investigated the effectiveness of Watts-type bright nickel and pyrophosphate copper (both electroplated coatings) as undercoats in minimizing the embrittlement induced by overplating with cyanide cadmium or conventional chromium electroplates. Both undercoats resulted in some decrease in the susceptibility to delayed failure under sustained load when the undercoats were thick enough. Since other workers purportedly have shown recently that neither hydrogen atoms nor protons diffuse through thin nickel layers, Beck and Jankowsky concluded that the beneficial effect of nickel and other metallic undercoats is the result of their action as diffusion barriers, and barrier effectiveness apparently is related to the thickness of the undercoat. They suggested that deficiencies in effectiveness of the barrier coating, as revealed by the sensitive delayedfailure test, is the result of plating imperfections.

The present authors have observed that a nickel flash (applied chemically to the entrance surface), such as is used in preparing a surface for vitreous enameling, resulted in very low, but measurable hydrogen permeation rates. However, when the nickel-flashed specimen was baked for several hours at 375 F prior to the permeation experiment, rather high permeation rates were obtained through the enameling iron.

Other Means of Hydrogen Entry

High hydrogen concentrations can be obtained in steel from the various reactions listed above. In addition, there are several other ways in which hydrogen can enter steel.

Exposure of steel to high-pressure hydrogen, especially at slightly elevated temperatures, or heating steel in hydrogen at lower pressures but at high temperatures introduces hydrogen into the steel because some dissociation of the molecular hydrogen occurs under these conditions.

In pressure charging, the pressure of hydrogen is the driving force for hydrogen entry into steel. In general, the rate at which a gas passes through a diaphragm is greater, the greater the pressure on the inlet side or the greater the difference in pressure between the inlet and outlet sides. Since gases dissolve in metals as atoms, not molecules, the rate is expected to be proportional to the square root of the pressure (at least for pressures low enough that the ideal gas laws are obeyed). This is well established for hydrogen in iron at elevated temperatures, as has been discussed previously. Hill and Johnson, (13) using the pressure-charging method, obtained solubilities of hydrogen in *a*-iron at 100-atmospheres pressure of hydrogen gas as follows:

Temperature		100-Atmosphere
С	F	Solubility, ppm
585	1085	10.98
538	1000	8.08
490	914	6.13
390	7,34	3.44
290	554	3.19
210	410	2.36
145	293	1.57

Other investigators recently reported the embrittlement of heat-treated 0.7 percent carbon steel by an external hydrogen atmosphere at room temperature. (149) Exposure of bars to hydrogen at 1,990 psi hardly affected the tensile strength, but the reduction in area became almost nil and the notch tensile strength (0.0028-inch notch radius) was reduced by 50 percent.

Actually in pressure charging and in thermal charging in a hydrogen-bearing atmosphere as well, the driving force is the partial pressure of dissociated (atomic) hydrogen. Hydrogen dissociates according to the reaction, $H_2 \longrightarrow 2H$, and it is the dissociated part of the gas that controls its solubility in metals. Because this dissociation produces an increase in volume, the reaction is pressure sensitive; increased pressure reduces the degree of dissociation. In effect, Sievert's law states that although the total number of hydrogen molecules present in a given volume increases directly as the pressure, the number of dissociated atoms increases only as the square root of the pressure.

Relatively little work has been done to determine the effect of elevated pressures on the permeation of hydrogen through steel, through its effect on the entry process. Conflicting results are found in the information that has been published. Baukloh and Guthmann(150) found that the hydrogen permeation rate at constant temperature for various steels increased linearly with the square root of the hydrogen partial pressure up to 40 to 100 atmospheres, and then it became virtually constant. At low pressures, well below 1 atmosphere, the results departed from the linear relationship, as has been observed by most investigators. On the other hand, Naumann(151) investigated the permeation of hydrogen through several steels at pressures up to 1000 atmospheres over a wide temperature range and found a linear relationship between permeation rate and the square root of the pressure up to the highest pressure investigated (except for the deviation at very low pressures).

Smithells and Ransley⁽⁷¹⁾ studied the "diffusion" of hydrogen through iron as a function of pressure (pressure charging in the pressure range from 0.094 to 142 mm Hg) at two temperatures, 775 and 1339 F (686 and 999 K). Their plot of rate of "diffusion" as a function of the square root of the pressure is shown in Figure 23. Although they called the effect diffusion, they actually were considering permeation and not diffusion through the bulk material, since surface variations markedly affected the results under some conditions. Their permeation values represented the volume of gas in cm³ at STP passing per second through 1 sq cm of surface of 1-mm thickness. At the higher pressures in the range studied, the observed data lay on straight lines; however, there was a definite departure from this relation at low pressures, where the rate of permeation was less than that required by the Richardson equation for permeation, P:

$$P = k / p_{\bullet}$$
(28)

The straight-line portions of the curves, if extended, cut the \sqrt{p} axis at small positive values corresponding to the "threshold" values (below which diffusion did not occur) described previously by Borelius and Lindblom. ⁽⁷²⁾ However, Smithells and Ransley found that this threshold does not exist because some permeation was observed at lower pressures. According to these investigators, all published data at the time exhibited the same deviation.

Smithells and Ransley then studied the effect on hydrogen entry of gas adsorbed on the surface of the iron. They considered that at low pressures







(Note change in the ordinate scales for the two figures by a factor of 10.)

and at temperatures where "diffusion" (permeation) could be measured, the amount of adsorbed gas is usually only a small fraction of a unimolecular layer. Therefore, at any given instant, hydrogen can enter only through those parts of the surface that are covered by adsorbed gas, the <u>effective</u> area of the metal surface being Θx , where x is the total area of the surface and θ is the fraction of the surface covered by adsorbed hydrogen. They introduced the concept that as the pressure, p, increased, θ increased according to the equation

$$\theta = \frac{ap}{1 + ap} , \qquad (29)$$

where a is a constant for a gas-metal system. Thus, at low pressures, θ increases rapidly and at high pressures it approaches unity. Introducing this adsorption isotherm into the permeation equation (for constant temperature) gave

$$P = k \sqrt{p(\frac{ap}{1+ap})}.$$
 (30)

At high pressures, the term in parentheses approaches unity and the equation approximates Equation 28. According to this equation, the rate of permeation would fall off at low pressures in the manner observed experimentally by a number of workers. Published permeation-rate data for several metals were plotted against the square root of pressure. Then, the curve for Equation 30 was drawn in after selecting suitable values for the constants a and k. The results Borelius and Lindblom obtained for iron at four temperatures have been treated in this way in Figure 24. The experimental points of Borelius and Lindblom are well represented by Smithells and Ransley's Equation 30.

Bryan and Dodge⁽⁴⁹⁾ studied the permeation rate of hydrogen in a highly purified iron (Ferrovac E) by maintaining a constant pressure of hydrogen inside a tubular specimen placed in an evacuated jacket and measuring the increase of pressure in the jacket. Permeation was studied for various pressures over the range from 1/30 to 300 atmospheres and for temperatures over the range from 259 to 1279 F (126 to 693 C). Although the generally accepted equations for solubility, diffusivity, and permeability indicate that all three properties vary as the square root of the pressure, Bryan and Dodge presented evidence that at high pressures the pressure should be replaced by the fugacity. They found that the permeation rate was linear with the square root of the fugacity rather than of the pressure, although the difference was not very great at the maximum pressure they used. These results are shown in Figures 25 and 26.

The dissociation of hydrogen is strongly endothermic. Therefore, dissociation is promoted by high temperatures, and this is the basis of thermal charging. Giauque(152) used spectrographic data to calculate the dissociation values of hydrogen for various temperatures, shown in Table 9. Although the dissociation is very small at room temperature, it becomes appreciable at steelmaking temperatures (about 2900 F). In the electric arc, hydrogen may be highly dissociated. Thermal charging is more effective if it is carried out in wet rather than dry hydrogen.

Both pressure charging and thermal charging with hydrogen have been discussed in more detail in another report dealing with the effects of hydrogen on steel at elevated temperatures and pressures (DMIC Report No. 202).

For some metals, hydrogen has been introduced by cyclotron proton irradiation. Exposure

/200









FIGURE 25. LOGARITHMIC PLOT OF PERMEA-TION RATE VERSUS PRESSURE AND FUGACITY, RUNS 103-121⁽⁴⁹⁾





TABLE 9. DISSOCIATION CONSTANTS FOR $H_2 = 2H (AFTER GIAUQUE)^{(152)(a)}$

Temperature			$K = \frac{(P_H)^2}{P_{H_2}}$	Percent Dissociation
K	С	F	P _{H2}	at l Atmosphere
298	25	77	$(1.3 + 5.6) \times 10^{-71}$	$(1.8 + 2.4) \times 10^{-34}$
500	225	435	$(7.7 + 13.5) \times 10^{-41}$ - 4.9	$(4.4 + 2.9) \times 10^{-19}$ - 1.7
1000	725	1335	$(6.5 + 4.3) \times 10^{-18}$	$(1.3 + 0.4) \times 10^{-7}$
1500	1225	2235	$(3.6 + 1.4) \times 10^{-10}$	$(9.5 + 1.7) \times 10^{-4}$
2000	1725	3135	$(2.97 + 0.85) \times 10^{-6}$ - 0.66	0.086 + 0.011 - 0.010
2500	2225	4035	$(6.93 + 1.54) \times 10^{-4}$ - 1.27 × 10 ⁻⁴	1.32 + 0.14 - 0.13
3000	2725	4935	$(2.66 + 0.49) \times 10^{-2}$ - 0.41 × 10 ⁻²	8.1 + 0.7 - 0.6
3500	3225	5830	$(3.68 + 0.57) \times 10^{-1}$ - 0.49 × 10 ⁻¹	29.0 + 2.0 - 1.8
4000	3725	6740	2.68 + 0.36 - 0.32	63.3 + 2.4 - 2.4
5000	4725	7630	43.6 + 4.7 - 4.1	95.7 ^{+0.4} -0.4

(a) Giauque made an error in calculating the precision of both the values of the equilibrium constant and the percent dissociation. The present authors have recalculated the values representing the precision, using Giauque's tabulated data, and these new values have been used in this table. to suitable flames is another method. Also, hydrogen has been charged into iron by glow discharge from a hydrogen atmosphere ionized by a directcurrent potential. (153,154,155)

Also, especially for heavy sections, hydrogen introduced during steelmaking operations may be a factor in delayed failures unless suitable vacuum processing has been used to minimize hydrogen pickup from such sources as water in the charge and in the refractories, including the ladle lining. Other sources of hydrogen during steelmaking are the scrap, hot metal, ferroalloys, and slag-making constituents. In fuel-fired melting furnaces, such as the open-hearth furnace, hydrogen and hydrocarbon gases usually are present in the furnace atmosphere. However, the most common source of hydrogen in liquid steel is water vapor which comes in contact with the liquid steel after it has been deoxidized. (156) This water vapor can originate in the air, furnace atmosphere, refractories, and slags. Thus, appreciable quantities of hydrogen are present in virtually all steel poured unless it is properly vacuum processed. Measures taken to minimize hydrogen defects include drying of raw materials (including slagmaking materials); careful drying of refractories in runners, ladles, and modes; and slow cooling and annealing of semifinished steel products. The use of oxygen during melting helps eliminate hydrogen. Today, considerable quantities of steel for high-strength applications are melted by the consumable-electrode vacuum-melting process, and these steels are virtually free of hydrogen in the ingot stage. Also, steel in thin sections loses hydrogen at an appreciable rate upon storage, so long as additional hydrogen is not being introduced by a corrosion process. However, hydrogen which enters the steel during melting can be held effectively in massive sections because the diffusion path to the surface is long. The hydrogen content in liquid low-alloy steel at the time of tapping may range from less than 1 ppm to at most 10 to 12 ppm; frequently the content is 4 to 6 ppm. Using the procedures listed above, the amount of hydrogen present may be limited to the extent deemed necessary.

Hydrogen also can be introduced into steel during heat-treating operations, although it is more likely that solid steel will absorb hydrogen from the steam reaction (Fe + H₂O \longrightarrow FeO + 2H) than from dissociation of molecular hydrogen gas.

For careful processing to minimize the hydrogen content of steel, the heat-treating atmosphere should not be conducive to hydrogen pickup from the steam reaction. Therefore, the watervapor content should be low; also, the partial pressure of hydrogen should be low. It will be remembered that treatment at elevated temperatures in a wet-hydrogen atmosphere is one method used to intentionally introduce hydrogen into steel. Frequently, a dry argon atmosphere is used in laboratory investigations. No acid pickling, cathodic cleaning, or electroplating operations can be tolerated, as these operations usually introduce hydrogen into steel. Vapor honing introduces more hydrogen into steel than does dry honing. Abrasion can introduce hydrogen into steel from water vapor in the air or from a cutting fluid. Also, the part must be protected from corrosion, including protection from rusting in moist air.

Effects of Surface Treatments and Environment

If the entry of hydrogen, rather than diffusion, is the controlling factor in permeation of hydrogen through steel, as appears to be the case under most conditions, then variations in the nature of the steel surface will influence hydrogen permeation.

It is well known that the specific surface of an adsorbent is usually larger than the measured projected surface; early estimates indicated that the two areas differ by a factor in the range of 1.5 to 20. Roughening the surface by etching or by alternate oxidation and reduction increases the area available for adsorption and also increases the proportion of the surface that is active. Thus, such treatments should affect the rate of permeation of hydrogen through steel at low pressures, and experiments have shown that this is so.

Smithells and Ransley(71) early studied the effects of such treatments on iron and nickel, and the results for iron are summarized as follows:

Temp K	erature F	Hydrogen Pressure, mm	Hydrogen Permeation Rate, cm ³ /sec/cm ² (a)
673	752	0.77	0.47×10^{-7}
673	752	0.77	4.4×10^{-7}
863	1094	0.073	1.28×10^{-7}
863	1094	0.073	0.76×10^{-7}
863	1094	0.073	1.54×10^{-7}
	<u>K</u> 673 673 863 863	673 752 673 752 863 1094 863 1094	Temperature Pressure, mm K F mm 673 752 0.77 673 752 0.77 863 1094 0.073 863 1094 0.073

(a) For thickness of 1 mm.

The oxidation was accomplished by introducing a small amount of oxygen into the permeation apparatus and heating the tubular specimen to form a thin oxide film. In the case of iron, great difficulty was experienced in removing the adsorbed oxygen, even when the specimen was heated in hydrogen at 1110 F (600 C), and the adsorbed oxygen reduced the rate of permeation. Such an oxidation-andreduction treatment doubled the hydrogenpermeation rate of nickel as compared to a polished surface. Removal of the adsorbed oxygen from the iron by a special treatment at 1470 F (800 C) resulted in a higher permeation rate than was obtained for the polished tube. Etching a polished surface with acid had a far more marked effect, increasing the permeation rate by a factor of 10.

That this etching increased the absorption of hydrogen into the specimen was indicated by a much larger value of the constant a in Equation 30 that was obtained for etched surfaces. This behavior also is shown by comparing Figures 23 and 24.

The present authors have observed that variations in the mechanical preparation of the surface (for example, grit blasting, vapor blasting, surface grinding, mechanical polishing, or electropolishing) can affect the hydrogen permeation rate considerably. At least a portion of this effect is undoubtedly the result of variations in surface roughness and, hence, gross surface area. However, the data obtained suggest that other factors also are involved, as the hydrogen entry rate was not always directly related to surface roughness.

Surface treatments such as activation or poisoning that are accomplished by exposure to suitable environments are known to affect adsorption, and hence hydrogen entry. Consequently, they affect hydrogen permeation, too. As was discussed in detail in a previous section, chemisorbed films of inhibitors or cathodic poisons can greatly reduce or increase the rate of hydrogen entry, respectively. Inasmuch as the effects of cathodic poisons, the nature of the electrolyte, and variations in current density are of major importance in the introduction of hydrogen into steel by cathodic charging, they have been discussed previously in the section dealing with cathodic charging. The effect of the environment on the entry of hydrogen also was discussed in the sections dealing with acid pickling and with other corrosion processes. The presence of moisture often is a factor; for example, thermal charging with hydrogen is much more effective when it is carried out in wet rather than dry hydrogen.

Recent laboratory work by Hudson and coworkers⁽⁸¹⁾ concerning the influence of oxidizing and other environments on the rate of desorption of hydrogen from steel that had been precharged with hydrogen showed that environmental conditions have considerable influence on the rate of desorption, and hence, on the permeation rate as well. This will be discussed more fully in a subsequent section dealing with the removal of hydrogen from steel.

Cold work can have a large effect on hydrogen permeation through steel. One might expect that part of the differences in permeation rate observed for different methods of mechanical surface treatment would be the result of differences in the depth of the thin cold-worked layer that most of them produce at the steel surface. However, since the effect of cold work on hydrogen permeation through iron and steel results primarily from its influence on hydrogen diffusion and the trapping of hydrogen inside the steel, rather than an effect on hydrogen entry, it will be discussed in the next section of this report, the one dealing with diffusion.

Diffusion

Most of the current theories of hydrogen embrittlement and hydrogen-induced delayed brittle failure of steel depend on diffusion of hydrogen in one way or another. Proper testing of these theories can only be accomplished when the diffusion process itself is properly understood. Although much work has been done recently on the diffusion of hydrogen in steel, the phenomenon still is not completely understood, and certain apparent discrepancies remain to be resolved. Unfortunately, as far as an understanding of delayed hydrogen cracking of steel is concerned, most of the serious discrepancies have been encountered at the lower temperatures in the vicinity of room temperature. The fact that relatively good agreement is beginning to be achieved for measurements in the general vicinity of room temperature would seem to indicate that there is hope of being able to treat the problem of hydrogen in steel from a scientific standpoint before long.

As was discussed previously, the entry and diffusion of hydrogen in metals occurs as atoms dissociated from the molecule, or perhaps as positively charged ions (protons) as suggested by Barrer(157) or partially shielded protons as suggested by Bastien. (8) In any event, molecular hydrogen (H2) does not diffuse in metals, nor do the molecules of any of the other diatomic gases. The chief evidence supporting this conclusion is the experimental finding that the rate of permeation of a diatomic gas through a metal under steadystate conditions generally is proportional to the square root of the pressure. This would be expected if the diffusion through the bulk solid were due solely to the movement of atoms, as follows from the equilibrium relationship

$$K = \frac{P_{H}}{\sqrt{P_{H_2}}},$$
 (31)

where K is the equilibrium constant of the dissociation reaction $(\frac{1}{2}H_2 = H)$, and p_H and p_{H_2} are the partial pressures of atomic and molecular hydrogen, respectively. The experimentally demonstrated exponential relationship between permeation and temperature also indicates the importance of dissociation into atoms. This is the effect one would expect if the rate-controlling process were an activated adsorption on the surface that resulted in some dissociation. Additional evidence is the observation that nascent hydrogen will diffuse in a metal against a high pressure of molecular hydrogen.

Apparently St. Claire Deville and Troost in 1863 were the first to observe that hydrogen diffuses through solid steel. The diffusivity of hydrogen depends upon the temperature, the nature of the steel, and the amount of hydrogen present in the steel at the time. The driving force for diffusion of hydrogen in steel can be either a hydrogenconcentration gradient or a stress gradient. Most diffusion processes occur in response to a concentration gradient. However, it was shown in studies of hydrogen-induced delayed brittle failure of steel that hydrogen also can diffuse in response to a stress gradient. For example, in a region of uniform hydrogen content, hydrogen can diffuse to the region of maximum stress just in front of a crack tip in a part under load. This has been discussed more fully in DMIC Report No. 196.

Effect of Temperature

The values of permeability, solubility, and diffusivity are consistent with the relationship $P = S \cdot D$ at elevated temperatures, but discrepancies arise at lower temperatures. Also, what appear to be anomalously low values of diffusivity are obtained at the lower temperatures, so diffusion in the two temperature ranges will be discussed separately.

<u>Elevated Temperatures</u>. The effect of temperature on the rate of diffusion can generally be expressed by

$$D = k \cdot e^{-Q/RT}, \qquad (32)$$

where T is the absolute temperature in degrees Kelvin, Q is the activation energy (calories per gram mole) necessary for diffusion, R is the gas constant (1.987 for the units given), and k is a constant. With these units, the diffusion coefficient D has the units cm^2/sec . Since this theoretical equation linking diffusivity and absolute temperature is exponential in form, a plot of $log_{10}D$ against $1/T_{abs}$ (Arrhenius-type plot) should be linear. This follows from the alternate form of Equation 32, a form which also is often used to express the results of diffusion experiments:

$$\log D = -\frac{Q}{4.574 \text{ T}} + \text{Const.}$$
 (33)

Bennek and Klotzbach⁽¹⁵⁸⁾ found a linear relationship between 1/T and the log of the permeability down to 570 F (300 C) for several steels. Hobson⁽¹⁵⁹⁾ found no anomaly in the permeability of a 3Cr-Mo steel down to 285 F (140 C). In other work, using specimens cut from large forgings, Hobson⁽¹⁶⁰⁾ showed that the evolution of hydrogen did not follow the predictions of the accepted exponential equation for diffusion at temperatures below about 300 F (150 C). Discontinuities in plots of hydrogen permeation through a thin diaphragm versus temperature have been observed by several workers. For example, Ham and Rast, (40) working with nearly pure iron, found anomalously low permeation at temperatures of 243 to 600 F (117 to 315 C). Chang and Bennett, (41) studying iron alloyed with chromium, nickel, or molybdenum, found anomalous permeability developing slowly at temperatures below 750 F (400 C) in Armco ingot iron and ironnickel alloys; they attributed this to a change in the mechanism of permeation, the rate being controlled by diffusion at higher temperatures but by surface processes (entry) as the temperature fell to the

570 to 750 F (300 to 400 C) range. On the other hand, an Arrhenius-type plot (the log of the permeability plotted versus the reciprocal of the absolute temperature) of the results of the permeation experiments performed by Bryan and Dodge(49) yielded an excellent straight line throughout the temperature range studied, as is shown in Figure 27. There was no evidence of a change in slope of the plot down to the lowest temperature that they studied (259 F or 126 C). Their equation for the temperature dependence of the permeability of hydrogen through pure iron was

$$\omega$$
 (or P) = 0.0227e^{-8120/RT}, (34)

where R is in gram-calories per gram-mole of hydrogen and T is degrees Kelvin.



FIGURE 27. ARRHENIUS PLOT SHOWING VARIA-TION OF PERMEABILITY WITH TEMPERATURE⁽⁴⁹⁾

Using Smithells and Ransley's permeability results(71) and Sieverts' solubility results, (32, 33)

Sykes, Burton, and Gegg⁽¹⁶¹⁾ calculated the diffusivity, D, for α -iron. The results, listed in Table 10, are compared with the permeability results obtained by Bennek and Klotzbach for a Ni-Cr-W steel. It is seen that the change in diffusivity of hydrogen in α -iron with change in temperature is relatively small. Over the temperature range from 570 to 1110 F (300 to 600 C), the diffusivity changed by a factor of 2 while the permeability changed by a factor of about 20. Thus, the major factor in improving the permeability is the increase in solubility with increasing temperature.

TABLE 10. COMPARISON OF DIFFUSIVITY AND PERMEABILITY FOR IRON IN THE α -STATE⁽¹⁶¹⁾

Temperature		Diffusivity, D,	Permeability	
С	F	sq cm/sec	N(a)	
20	68	1.5×10^{-5}		
100	212	3.5×10^{-5}	0.00026	
200	390	6.7×10^{-5}	0.0045	
300	570	1.00×10^{-4}	0.029	
400	750	1.38×10^{-4}	0.11	
500	930	1.70×10^{-4}	0.26	
600	1110	2.04×10^{-4}	0.59	
700	1290	2.34 x 10^{-4}	1.00	
800	1470	2.69×10^{-4}		

(a) N is the number of cc of gas at NTP diffusing through 1 sq cm of a plate 1 mm thick in 1 hr at a pressure of 1 atm, as determined by Bennek and Klotzbach. N corresponds to Sykes, Burton, and Gegg's Z and to P used by most investigators.

Geller and Sun⁽³⁸⁾ determined the diffusion coefficient for high-purity α -iron at elevated temperatures. Their expression for hydrogen diffusion was

$$Log D = -\frac{637}{T} - 2.663,$$
 (35)

where D is the diffusion coefficient with units of cm^2/sec and T is degrees Kelvin. To facilitate comparisons to be made later, this equation can be rewritten as follows:

$$D = 0.0022e^{-2914/RT}.$$
 (36)

More recently, Stross and Tompkins⁽¹⁶²⁾ determined the diffusion coefficient of hydrogen in α -iron for the temperature range from 300 to 1650 F (150 to 900 C) from measurements of the rate of evolution into a vacuum from an iron cylinder saturated with hydrogen under various conditions. They found that at temperatures above 300 F (150 C) diffusivity in α -iron conforms to the relationship

$$D_{(cm^2/sec)} = 8.8 \times 10^{-4} e^{-3050/RT}$$
. (37)

Essentially the same values of D have been determined by Eichenauer, Kunzig, and Pebler(163) and by Johnson and Hill. (10) The latter workers studied the diffusivity in α -iron by measuring evolution rates over the temperature range from 77 to 1435 F (25 to 780 C). Irons were prepared by air melting and by vacuum melting, and both pressure-charging and thermal-charging methods were used to introduce hydrogen. The different melting and charging methods gave comparable results, as is shown in Figure 28. Above 392 F (200 C), the diffusivity, D, in cm²/sec was represented by

$$D = 0.0014e^{-3200/RT}$$
. (38)

However, below 392 F (200 C), the diffusivity was anomalously low, as will be discussed in the next section.



FIGURE 28. DIFFUSION COEFFICIENT OF HYDRO-GEN IN IRON⁽¹⁰⁾

As was discussed previously, the permeability, solubility, and diffusivity generally are related as follows:

$$P = S \cdot D$$
, (12)

(also see Equation 8). This is the relationship that Sykes et al⁽¹⁶¹⁾ used to calculate the diffusivity from the permeability data for α -iron obtained by Smithells and Ransley⁽⁷¹⁾ and solubility data obtained by Sieverts.^(32,33) Their results for the diffusivity constant for α -iron, expressed as cm²/sec, were as follows:
$D = 7.6 \times 10^{-4} e^{-2285/RT}$ (39)

This equation agrees reasonably well with that which Stross and Tompkins subsequently determined experimentally (see Equation 37). This agreement indicates that the measurements of these three properties (permeability, solubility, and diffusivity) are individually reliable, and that the results are mutually consistent and in agreement with the relationship $P = S \cdot D$ for temperatures above about 750 F (400 C).

Lower Temperatures (Below About 750 F). Although a number of investigations have been made of the diffusivity of hydrogen in iron and steel at room temperature and slightly elevated temperatures, the various values obtained are in poor agreement. For example, for five investigations, values reported for the diffusion coefficient at 77 F (25 C) ranged from about 1×10^{-9} to 1×10^{-5} cm²/ sec; these values differ by a factor of 10,000. However, the various results are in accord in one respect -- all follow the trend for experimentally determined coefficients for temperatures below about 390 F (200 C) to be considerably lower than would be predicted by extrapolation of high-temperature measurements for α -iron. This observed behavior of hydrogen in ferrite is just the reverse of its behavior in most other metal systems where lowtemperature deviations in the diffusion coefficient have been in the direction of higher values than are predicted by extrapolation of high-temperature data. These deviations in metals other than iron or iron alloys have been explained as resulting from preferentially rapid diffusion along grain boundaries or dislocations.

Because considerable controversy existed as to whether hydrogen diffuses through the lattice of metal crystals, through the grain boundaries, or through faults in the crystals, Smithells and Ransley⁽⁷¹⁾ conducted an experiment to obtain direct evidence on the effect of grain boundaries. They performed diffusion measurements on a singlecrystal tube and on another piece of the same tube of iron treated so as to have a normal grain size of 100 grains per mm². The effect of grain size on the rate of diffusion of hydrogen through iron is shown in the following tabulation:

Temperature Pr		e, Diffusi	Diffusion Rate(a)		
F	mm	Fine Grain	Single Crystal		
473	140	2.4×10^{-6}	1.2 x 10-6		
775	140		17.1×10^{-6}		
1150	140	92.8 x 10 ⁻⁶	89.5 x 10-6		
1434	140	203.0×10^{-6}	205.0×10^{-6}		
	F 473 775 1150	F mm 473 140 775 140 1150 140	F mm Fine Grain 473 140 2.4 x 10 ⁻⁶ 775 140 17.6 x 10 ⁻⁶ 1150 140 92.8 x 10 ⁻⁶		

(a) Units not stated.

The results show that the presence or absence of grain boundaries had no appreciable effect on the rate of diffusion of hydrogen through iron. Ed-wards (164) also had found the diffusion rates to be the same in a single crystal as in the same material

in fine-grain polycrystalline form. Thus, it is accepted that the penetration of hydrogen into iron occurs throughout the lattice.

Using specimens cut from large forgings of 2-1/2Ni-Cr-Mo steel, Hobson⁽¹⁶⁰⁾ showed that the diffusion of hydrogen in steel does not follow the predictions of the accepted exponential equations for α -iron (for example, see Equations 39 and 37, derived from data obtained at elevated temperatures) at temperatures below about 300 F (150 C). Instead, the diffusion rate and the diffusion coefficient are anomalously low at the lower temperatures, as is shown in Figure 29 and in the following tabulation:

Tempe	rature	Diffusion Rate Relative to RT	Relative Rate Predicted by Sykes-Burton- Gegg Equation	Observed Diffusion Constant,
С	F	Taken as Unity	(Eq. 39)	cm ² /sec
- 78	-108	0.018	0.14	5.8 x 10-9
RT	RT	1	1	3.2 x 10-7
50	122	4.2		1.35 x 10-6
100	212	41.9	2.3	1.34×10^{-5}
200	392	209	4.5	6.7 x 10 ⁻⁵



FIGURE 29. OBSERVED AND CALCULATED RATES OF DIFFUSION OF HYDROGEN IN IRON(160)

[Showing anomalously low diffusion of hydrogen in alpha iron at low temperatures - below about 390 F (200 C) in this instance.]

Hobson suggested that the diffusion experiments revealing anomalously low rates of hydrogen diffusion in steel below about 390 F (200 C) and a great reduction in diffusion rate below 212 F (100 C) (see Figure 29) make the continuation of slow cooling below the latter temperature of little value in reducing the hydrogen content of large forgings. Yet numerous investigators in sampling steel for hydrogen analysis have shown that for small specimens serious losses occur at room temperature.

Darken and Smith⁽⁴⁸⁾ measured the effect of temperature on the permeability of hydrogen introduced into SAE 1020 steel by acid charging. The results may be summarized as follows:

Permeability at Indicated Temperature,

	cm ²	/sec/cm ⁻ for	l mm thickne	SS
Solution	32 F (0 C)	95 F (35 C)	131 F (55 C)	167 F (75 C)
$1N H_2SO_4$	5.2 x 10-6	23×10^{-6}		78 x 10-6
l molal citric acid		6 x 10 ⁻⁶		34 x 10-6
pH 4.4		1.04×10^{-6}	1.31 x 10-6	1.46×10^{-6}

The permeability doubled for about 36 F (20 C) rise in temperature (except for the pH 4.4 solution where there was a large change in potential). Bardenheuer and Thanheiser(62) found about the same temperature coefficient for a steel exposed on one side to sulfuric acid (387 g/l), as follows:

Temperature		
С	F	Permeability x 10^{6}
20	68	2.03
30	86	2.56
40	104	3,93
50	122	5.16
60	140	9.1
70	158	11.8
80	176	18.6

Frank, Swets, and Fry⁽⁹¹⁾ recently determined the diffusion rates of hydrogen in mild steel at low temperatures (77 to 194 F or 25 to 90 C) where the diffusion rates are relatively low. A mass spectrometer was used to study the movement of hydrogen through thin steel plates of different thicknesses. One surface of the plate faced the inside of the mass spectrometer and hydrogen was introduced into the other (outside) surface. Hydrogen was introduced by abrasion of the steel under water, the most satisfactory of several methods tried. Two methods of measuring diffusion were used: (1) the rate of approach to equilibrium when hydrogen was supplied to the steel plate. and (2) the rate of outgassing when the source of hydrogen was removed. Two markedly different sets of diffusion coefficients were obtained by the two methods. The first method gave diffusion coefficients that agreed very well with values obtained by others at somewhat higher temperatures.

The outgassing method, which was the more reproducible method, gave considerably lower values, as shown in Figure 30. The results are compared with those of other investigators in Figure 31. For the curves shown in Figure 30, the equation which fits the upper curve ("build-up" method) is

$$D = 5.0 \times 10^{-3} e^{-3400/RT}, \quad (40)$$

and that for the lower curve (decay method) is

$$D = 1.90 \times 10^{-2} e^{-6320/RT}$$
, (41)

where the activation energy is in calories per mole and the value of D is $\rm cm^2/sec$. The decay method resulted in higher values of activation energy and higher D_o values (the constants before the "e" terms in the two equations) than did the build-up method.



FIGURE 30. PLOTS OF 1nD VERSUS 1/T x 1000 FOR THE DIFFUSION OF HYDROGEN IN MILD STEEL(91)

[For the "build-up" method (upper) and the decay or outgassing method (lower) applied to the annealed specimen.]

In the experiments that involved making measurements from the decay curves, the effect of concentration was investigated through doubling the maximum permeation rate by increasing the load on the abrasive paper in one case and increasing the concentration of citric acid in the other. No significant change was detected in the decay rate, and hence, none in the diffusivity. This indicated



FIGURE 31. A COMPARISON OF THE WORK OF FRANK, SWETS, AND FRY ON THE DIFFUSION OF HYDROGEN IN STEEL AT LOW TEMPERATURES WITH THE WORK OF OTHERS(91)

that if there was a concentration effect, it was at least very small compared to other effects. Two observations indicated that the difference between the two sets of diffusion coefficients was not the result of surface effects. First, there was no significant difference in diffusion coefficients obtained from plates of different thickness, a result which would not be expected if surface effects were controlling. Second, when extensive changes were made in the character of the surfaces, by acid attack and other means, only minor differences in the decay rate resulted; these differences were small compared to the difference between the two sets of diffusion coefficients. Also, the difference between the two sets of coefficients cannot be attributed to the abrasion process because the same effect was observed when hydrogen was introduced by immersion in citric acid. The authors listed several facts which tentatively point to a time effect. Part of these were: (1) the "build-up" process preceded the decay process in time and gave the higher diffusion coefficients, (2) if hydrogen introduction was stopped early, before steady-state conditions were achieved, the decay rate was faster, and (3) if the hydrogen-introducing process was maintained a longer time after steady state was reached, the decay rate was always slightly lower. It was suggested that the one set of diffusion coefficients (from "build-up" measurements) were possibly associated with the simple, interstitial diffusion process, while the other set, which appeared after the hydrogen had been in the steel for awhile,

apparently was associated with a more complex process. In a subsequent note, Frank⁽¹⁶⁵⁾ discussed a similar effect observed by Weiner and Gensamer⁽¹⁶⁶⁾ who used internal-friction measurements to study hydrogen in mild steel in the very low-temperature range (subzero). Weiner and Gensamer also found that there were two processes of hydrogen movement and that they involved different activation energies. Furthermore, the process involving the higher activation energy occurred only after the hydrogen had been in the steel for some time, just as observed by Frank and co-workers.

Frank then presented the summary of the status of measurements of the diffusion coefficients of hydrogen in iron and steel that is shown in the top part of Table 11. In this type of work, as he pointed out, one is concerned primarily with whether the diffusion coefficients obey the law

$$D = D_0 \cdot e^{-E/RT}, \qquad (42)$$

and if they do, one would like to know what the values of E and D_0 are, E being considered the activation energy of the diffusion process. The seven sets of measurements in the table cover the temperature range from -369 to 1650 F (-223 to 900 C); all were made on high-purity iron or annealed mild steel, and several different methods were used.

TABLE 11. SUMMARY OF STATUS OF MEASUREMENTS OF DIF-FUSION COEFFICIENTS OF HYDROGEN IN IRON AND STEEL

Temperat	ure Range	Do,	E,	
С	F	cm ² /sec	cal/mole	
150-900	300-1650	8.85±0.83 x 10 ⁻⁴	3050±100	
400-900	750-1650	4.6 x 10 ⁻²	2900	
100-800	210-1470	7.6 x 10^{-4}	2290	
25 - 90 25 - 90	77-195 77-195	5.0×10^{-3} 1.9×10^{-2}	3400 6300	
er -223 -168	-369 -270	4×10^{-4} 9 x 10^{-5}	3000 6000	
126-693	259-1279	3.9 × 10 ⁻⁴	1080	
200 -7 50 25 - 200	392-1382 77-392	1.4 x 10 ⁻³ 0.12	3200 7820	
	C 150-900 400-900 100-800 25-90 25-90 cr -223 -168 126-693 200-750	150-900 300-1650 400-900 750-1650 100-800 210-1470 25-90 77-195 25-90 77-195 -168 -270 126-693 259-1279 200-750 392-1382	C F cm ² /sec 150-900 300-1650 8.85 ± 0.83 × 10 ⁻⁴ 400-900 750-1650 4.6×10^{-2} 100-800 210-1470 7.6×10^{-4} 25-90 77-195 5.0×10^{-3} 25-90 77-195 1.9×10^{-2} acr -23 -369 4×10^{-4} -168 -270 9×10^{-5} $126-693$ $259-1279$ 3.9×10^{-4} 200-750 $392-1382$ 1.4×10^{-3} $302-1382$ 1.4×10^{-3}	

(a) This work was performed subsequent to Frank's summary.

In the upper temperature region, there appears to be only the interstitial diffusion process with an activation energy of approximately 3000 cal/mole. However, there appear to be two diffusion processes in the lower temperature range. When hydrogen is first introduced into steel, it moves around relatively freely, and the diffusion process has an activation energy of about 3000 cal/ mole. But, with the passage of time, the hydrogen

appears to be partially trapped by some lattice defect, and the diffusion of this hydrogen takes more energy to become activated (approximately 6000 cal/mole is required) and takes place at a lower rate. Although there are differences in the values of D_0 in the table, when one considers that they are extrapolations of the measured values to infinite temperature (that is, 1/T = 0) and thus are sensitive to inaccuracies in the measurements, they agree with each other amazingly well.

Other recent work on the diffusion of hydrogen in iron and in iron-carbon alloys has been that of Hill and Johnson. (10,14,13,167) All the results shown in Figure 32 were obtained from measurements of the nonsteady evolution of hydrogen that had been introduced into specimens by heating them in hydrogen gas at various pressures up to 150 atmospheres. The uppermost curve in the figure shows the diffusivity of hydrogen in annealed, pure α -iron at temperatures from 77 to 1380 F (25 to 750 C). Rather than showing the straight line one expects to find in this type of plot (log D versus 1/T), the curve shows a distinct break in the vicinity of 390 F (200 C). Above 390 F, the diffusivity, D, in cm²/sec was shown earlier in this report to be represented by

$$D = 0.0014e^{-3200/RT}$$
 (38)

However, below 390 F, the diffusivity was anomalously low, being represented by

$$D = 0.12e^{-7820/RT}$$
. (43)

At room temperature (77 F or 25 C), this yields a value of D of 2.2 x 10^{-7} cm²/sec.

As was dissussed in the preceding section, the results obtained above 390 F(200 C) agreed fairly well with those of other recent investigations. These results of Johnson and Hill⁽¹⁰⁾ were obtained with several heats of iron which varied in total impurity content between 0.02 and 0.1 percent. In the higher temperature range, measurements of D were found to be quite reproducible in spite of variations in impurity content, grain size, thermal and mechanical history, and original hydrogen content. The investigators were satisfied that surface processes were not controlling factors in the evolution of hydrogen and that the values reported were in fact volume-diffusion coefficients.

The anomalously low values of D that Johnson and Hill obtained for temperatures below 390 F (200 C) were not entirely unexpected because they are in agreement with the general behavior observed by some of the other recent investigators [for example, Schuetz and Robertson, (57) Ham and Rast, (40) Hobson, (160) and Frank, Swets, and Fry(91)]. The abnormally low diffusion rates found by Ham and Rast at low temperatures are illustrated in Figure 33. These anomalously low values are not fully understood. Johnson and Hill hypothesized that at low temperatures the hydrogen in the metal is "trapped" with an energy 4800 cal/g-atom below that of the interstitial hydrogen.







FIGURE 33. DIFFUSION ISOBAR [666-196 F (352-91 C)] AT ATMOSPHERIC PRESSURE FOR CARBONYL IRON BAKED 5 HOURS AT 2010 F (1100 C) IN H₂--"TRAPPED" METHOD(40)

(Rates are measured in g/sec through sample. Area = 2.8 cm^2 . Thickness = 0.155 cm.)

Bastien and co-workers have performed many investigations to learn more about the behavior of hydrogen in steel. Recently, Veysseyre, Azou, and Bastien⁽¹⁶⁸⁾ determined the diffusion coefficient, D, of hydrogen in iron at 77 F (25 C) by measuring the amount of hydrogen that passed into the interior of a hollow soft-iron tube which was the cathode during electrolysis in a N HCl solution. The hydrogen was drawn off under vacuum and its volume measured. They obtained a value of D of approximately 3.0×10^{-6} cm²/sec at this temperature.

The results of recent determinations of the diffusion coefficient at room temperature (77 F or 25 C) are compared in the following tabulation:

Investigators	D, cm ² /sec
Johnson and Hill	2.2×10^{-7}
Frank, Swets, and Fry ("build-up" method)	1.6×10^{-5}
Frank, Swets, and Fry ("decay" method)	4.5 x 10^{-7}
Veysseyre, Azou and Bastien	~3.0 x 10 ⁻⁶

Although the maximum and minimum values in the tabulation differ by a factor of 74, the agreement is encouraging, considering the much wider variation obtained in earlier work and the difficulties encountered in measuring diffusivity at room temperature.

On the other hand, Bryan and Dodge, (49) in studying the permeation of hydrogen through highly purified iron, found no break in an Arrhenius-type plot of log D versus 1/T when the temperature went below some temperature in the range of 750 to 390 F (400 to 200 C). Instead, their data point for 259 F (126 C) fit the straight line through the data points for higher temperatures very precisely. However, their activation energy was lower than other values reported in the literature. Also, Stross and Tompkins(162) carried their experiments down to 302 F (150 C) and found no change in slope of an Arrhenius-type plot of the results. Evidently, more work is needed to understand the reason for these differences from the observations of Johnson and Hill and of Frank, Swets, and Fry.

One of the most recent analyses of the diffusion of hydrogen in iron and ferritic steel to be reported in the literature is that of McNabb and Foster. (169) They reported that a consideration of the literature shows the inadequacy of Fick's laws to describe the diffusion of hydrogen in steels below 750 F (400 C). Therefore, McNabb and Foster modified these laws on the assumption that hydrogen is delayed at fixed sites in the lattice. In a highly mathematical treatment, approximate and exact solutions of the resulting nonlinear system were derived for simple geometrical shapes. The results purportedly account qualitatively for the observed anomalous behavior. Experimental methods for verifying the theory were suggested.

Anomalous decreases in permeation rate at low temperatures also have been found by several investigators, and it is believed that they are related to anomalous changes in the diffusivity.

Effect of Pressure

A number of investigators showed that, for elevated temperatures, the solubility of hydrogen in iron is proportional to the square root of the hydrogen partial pressure, as was discussed previously. Since the rate of diffusion (in the case where a concentration gradient is the driving force) depends on the concentration of gas in the metal, one might expect that diffusion also would be proportional to the square root of the gas pressure, p, according to the Richardson equation:

$$D = k / p.$$
 (44)

However, as early as 1912-1913, Charpy and Bonnerot showed that hydrogen diffuses in steel against a back-pressure of molecular hydrogen greater than 14 atmospheres. Bardenheuer and Thanheiser(62) found that diffusion was still taking place when the back-pressure of molecular hydrogen reached 300 atmospheres, the limit of their equipment. Theoretically, the equilibrium back-pressure at room temperature is far higher than this for ordinary hydrogen contents, as has been discussed previously in considering the effect of pressure on solubility.

Since the effect of hydrogen partial pressure on permeation is an effect on hydrogen entry rather than on diffusion in the bulk solid, it has been discussed previously in the section dealing with hydrogen entry into iron and steel.

Effect of Crystal Structure

Chang and Bennett⁽⁴¹⁾ determined the diffusion constant for hydrogen in an iron-4.36 percent nickel alloy in both the ferritic and the austenitic condition; the results are shown in Figure 34. Note that, for a given temperature, the diffusion constant for v-iron was much smaller than that obtained for α -iron.

Equations 35 and 36 give the results that Geller and Sun⁽³⁸⁾ obtained for the diffusion coefficients for high-purity iron in the α -form. The expression that they obtained for this same iron in the γ form was as follows:

$$Log D = -\frac{2174}{T} - 1.971.$$
 (45)

For ease of comparison, this equation can be rearranged as follows:

$$D = 0.0107 e^{-9944/RT}$$
 (46)





(Showing the effect of crystal structure on diffusion.)

These results are plotted in Figure 35, along with the results for α -iron (and for several iron alloys as well). This figure, too, shows that the diffusivity of hydrogen is markedly lower in γ -iron than it is in α -iron at the same temperature. In the case of α -iron, even though the hydrogen solubility is appreciably lower than it is in γ -iron at the same temperature, the passage of a hydrogen atom from one site in the iron lattice to another is much easier, and, therefore, the activation energy for diffusion in α -iron is lower.

Sykes, Burton, and Gegg⁽¹⁶¹⁾ calculated the diffusivity of hydrogen in austenite using the method described previously for ferrite. Sieverts' solubility data^(32,33) and Bennek and Klotzbach's data for permeability⁽¹⁵⁸⁾ were used for the austenitic state. The results of their calculations are shown in Table 12. If these results are compared with those obtained for the α -condition shown in Table 10, one finds that only when the temperature reaches about 2100 F (1150 C) does the diffusivity in γ -iron reach that determined for α -iron at 1200 F (650 C).

Hobson, (160) in studying diffusion in specimens cut from large forgings of an alloy steel, also obtained data that show that the diffusivity in austenite is much lower than in ferrite for a given temperature. Part of his findings are shown in Figure 29.

Bennek and Klotzbach(158) measured the permeation of hydrogen through 10 steels of widely



FIGURE 35. DIFFUSION COEFFICIENT OF HYDRO-GEN IN IRON AS A FUNCTION OF TEMPERATURE, CRYSTAL STRUC-TURE, AND ALLOY ADDITION⁽³⁸⁾

TABLE 12. DIFFUSIVITY AND PERMEABILITY OF HYDROGEN IN IRON IN THE y-STATE(161)

Temperature		Diffusivity, D;a	Permeability, Z,
С	F	sq cm/sec	volume/hr
1200	2190	2.5×10^{-4}	6.5 (calculated)
1100	2010	1.9×10^{-4}	3.9 (calculated)
1000	1830	1.3×10^{-4}	2.4
900	1650	8.7×10^{-5}	1.2
800	1470	6.0×10^{-5}	0.55

(a) There are not so many permeability data available for the γ -condition as for the α -condition, so the diffusivity figures may not be so reliable.

varying composition, over a wide temperature range (limits of about 570 and 2310 F or 300 and 1265 C) for a plate thickness of 1 mm. The compositions of their steels are listed in Table 13. The results obtained for four carbon steels of widely varying carbon content and five alloy steels, including 18Cr-8.9Ni austenitic stainless steel, are shown in Figure 36. Steel composition had some





FIGURE 36. HYDROGEN PERMEABILITY OF NINE STEELS INVESTIGATED AS A FUNCTION OF TEM-PERATURE FOR BOTH HEATING AND COOLING(158)

o Heating x Cooling 47

			Con	npositio	ition, percent			Hydrogen Dissolved at 1470 F
Steel	Steel C Si Mn Cr Mo Ni Other	Other	(950 C); cm ³ /100 g					
1	0.03	0.01	0.12					
2	0,13	0.14	0.35					5.64
3	0.47	0.26	0.71					
4	0.99	0.31	0.45					4.97
5	0.21	0.24	0.47	0.72		3.62		
6	0.13	0.30	0.40	1.52		4.11	1.03	W 5.56
7	1.06	0.42	0.30	1.53				5.61
8	0.25	0.24	0.90	2.68	0.24			5.70
9	0.10	0.67	0.40	18.0		8.9		7.60
10	0.24	0.28	0.90	2.72	0.26		0.52	A1 5.84
20		0.00	0.70		0.20			

TABLE 13. CHEMICAL COMPOSITION AND HYDROGEN SOLUBILITY OF THE STEELS INVESTIGATED BY BENNEK AND KLOTZBACH⁽¹⁵⁸⁾

influence on permeability, as will be discussed in the next section of this report. However, the most noteworthy result was the big drop in permeability that accompanied the α -to- γ transformation on heating, and the corresponding sharp increase for the reverse transformation on cooling. The data for the austenitic stainless steel are consistent with the data obtained for the other eight steels for temperatures at which they were in the austenitic state. The difference in permeability between the α - and Y-states for the same temperature is shown clearly in Figure 37 for the Cr-Ni-W steel (Steel 6) and the mild steel, where the curves have been extrapolated to lower temperatures. Numerically, the following values were obtained from the extrapolated curves for the Cr-Ni-W steel (Steel 6) for the amount of hydrogen passing through a membrane 1mm thick under a pressure differential of 760 mm of mercury:

Temperature		Permeation, N	$, cm^3/hr/cm^2$	
C F		In the Y-Region		
1000	1832	2.4		
900	1652	1.2		
800	1472	0.55		
700	1292	0.24	1.0	
600	1112	0.085	0.59	
500	932	0.018	0.26	
400	752	0.0032	0.11	
300	572	0.00027	0.029	
200	392	0.000008	0.0045	
100	212	0.00000034	0.00026	

Effect of Steel Composition

Steel composition, if it has an effect on the movement of hydrogen in steel, might be expected to affect the diffusion of hydrogen through the bulk solid and, thereby, affect permeation. A few investigations have attempted to show the effect of steel composition on the diffusivity of hydrogen in steel. Most of the rest have merely examined the effect on overall permeation rate.



FIGURE 37. COMPARISON OF THE HYDROGEN PERMEABILITY OF AUSTENITE AND FERRITE AT VARIOUS TEM-PERATURES (CURVES EXTRAPOLA-TED)⁽¹⁵⁸⁾

Geller and Sun(38) studied the effects of alloy additions upon the diffusivity of hydrogen in iron. The results, shown in Figure 35, indicate that alloy additions of silicon, chromium, or nickel decrease the diffusivity in either the α - or the γ modification of iron. In measurements at room temperature, these investigators observed much smaller hydrogen-evolution rates for a supersaturated alloy containing 4.33 percent silicon than for a 25.37 chromium-iron alloy. Specimens of alloys with 5.34 and 8.0 percent silicon (balance iron) showed no hydrogen evolution at all at room temperature.

Hobson(160) observed very low diffusion rates for a 14 percent chromium, ferritic stainless steel, as is shown by the following comparison:

		iffusion Constance, cm ² /sec	ant at Indicated
Steel	-108 F (-78 C)	Room Temp	212 F (100 C)
2½Ni-Cr-Mo	5.8 x 10 ⁻⁹	3.2 x 10-7	1.34 x 10 ⁻⁵
3Cr-Mo	$\sim 2.5 \times 10^{-9}$	$\sim 5.0 \times 10^{-8}$	
14Cr		~2.8 x 10 ⁻⁹	~3.2 x 10 ⁻⁷

It appears that the presence of chromium causes a large reduction in the rate of diffusion of hydrogen in iron. The high-chromium steels, though ferritic, are considered to be less susceptible to hydrogen defects than are chromium-free ferritic steels. This behavior may result from the much lower hydrogen-diffusion rates in the high-chromium steels as compared to steels low in chromium. Thus, one would expect the induction period before the onset of hydrogen cracking under sustained load to be much longer in high-chromium steels.

Chang and Bennett⁽⁴¹⁾ studied the effects of chromium, nickel, and molybdenum on the rate of hydrogen permeation in iron. They found that chromium had little effect on permeation in the Y-range, but greatly reduced the rate of permeation in the α -range. Nickel and molybdenum had little effect. Permeation in an Fe-4. 25Cr alloy was less than one-tenth the rate obtained for Armco ingot iron over the temperature range from 750 to 1380 F (400 to 750 C), as is indicated in Figures 38 and 39. Note the anomalously low permeation rates obtained for the ingot iron at the lower temperatures, a finding which is consistent with observations by other investigators and which reflects the anomalously low diffusion rates obtained in the same temperature range (for example, see Figures 29, 31, 32, and 33). The effect of chromium on permeation reflected the reduction in diffusivity by chromium.

Bhat and Lloyd(170) also observed that the addition of chromium effected a sizable reduction in permeation rate, as is shown in the following tabulation of the results of permeation during cathodic charging:

Steel C	Composi	tion, %		Hard- ness,	Time to Start of Evolu-	Rate of Permea- tion.
С	Ni	Cr	Treatment	Vickers	tion, min	cm ³ /hr
0.36	3, 28		Water quenched from 1560 F and tempered at 1200 F	245	42	0.42
0.32	3.29	0.87	Ditto	302	50	0.18



FIGURE 38. PERMEATION ISOBARS OF ARMCO IRON(41)



FIGURE 39. PERMEATION ISOBARS OF Fe-Cr ALLOYS(41) These investigators also determined the effect on permeation of variation in carbon content for normalized structures charged with hydrogen electrolytically. The results were as follows:

Carbon Content %		Hard- ness, Vickers	Time to Start of Evolu- tion, min	Rate of Permea- tion, <u>cm³/hr</u>
0.06	Normalized from 1705 F	98	18	0.44
0.31	Normalized from 1650 F	156	25	0.28
0.59	Normalized from 1560 F	202	39	0.22

Their results indicated that increasing the carbon content decreased the permeation rate at room temperature considerably. According to Smialowski, (119) all opinions are now consistent with the statement first made by Bardenheuer and Thanheiser in 1929 that the presence of carbon considerably diminishes the hydrogen-permeation rate at ordinary temperatures. However, as will be shown subsequently in this report, the structural form of cementite (Fe₃C) greatly influences the permeability of hydrogen through steel. At high temperatures, the effect of carbon content on hydrogen permeation through steel appears to be just the reverse of the above behavior, as is indicated by the results of Lewkonja and Baukloh's⁽¹⁷¹⁾ experiments that are summarized in Figures 40 and 41. At the temperatures they used, hydrogen attack produces decarburization in time, and this phenomenon could have complicated the experiment.

As was discussed in the section dealing with the effect of crystal structure, Bennek and Klotzbach⁽¹⁵⁸⁾ measured the permeation of hydrogen through 10 steels of widely varying composition over a wide temperature range. The steels investigated were listed in Table 13. Steel composition had some influence on the permeability of hydrogen



FIGURE 40. HYDROGEN PERMEATION OF PLAIN-CARBON STEELS AT 1562 F (850 C) AS A FUNCTION OF TIME (5-MM WALL THICKNESS)⁽¹⁷¹⁾

in a-iron, as can be seen from Figure 36. For the temperature range from 1380 to 750 F (750 to 400 C), the results they reported for eight of the nine ferritic steels (including four carbon steels with 0.03, 0.13, 0.47, and 0.99 percent C, and four alloy steels including 0.72Cr-3.62Ni, 1.52Cr-4.11Ni-1.03W, 1.53Cr, and 2.68Cr \cdot 0.24Mo) gave an average value of 5000 \pm 300 for the constant B in Equation 11 (page 16). Thus, the alloy additions had little if any effect on the activation energy for permeation (B·R) and, consequently, on the activation energy for diffusion, since these experiments were performed at elevated temperatures where diffusion was rate controlling.

Effect of Steel Microstructure

Not only does the presence of carbon have considerable influence on the permeation of hydrogen through iron or steel, as was discussed in the preceding section of this report, but the physical form of the carbide particles exerts a very considerable influence, too. This effect arises largely from the effect of microstructure on diffusivity. Bardenheuer and Thanheiser(62) investigated the effect of microstructure in a study of the acid pickling of steels that contained from 0.7 to 1.3 percent carbon. They observed 12 times greater hydrogen permeation rates when the cementite (iron carbide) was in globular form than when it was in the form of coarse lamellar pearlite. They also found that quenched steels had a lower permeability than annealed ones exhibited.

Smialowski⁽¹¹⁹⁾ has described experiments that Nosyreva reported in 1948 in which the structure dependence of hydrogen permeation through a 0.84 percent carbon steel was studied. Nosyreva compared the permeation rates of cathodically evolved hydrogen at room temperature after various thermal treatments. The results of these experiments are given in Table 14. Nosyreva obtained



FIGURE 41. HYDROGEN PERMEATION OF PLAIN-CARBON STEELS AT VARIOUS TEM-PERATURES AS A FUNCTION OF CARBON CONTENT⁽¹⁷¹⁾

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a maximum permeability for the normalized condition. Bhat and Lloyd. (170) using electrolytic charging, compared the hydrogen permeation rates of various structures for a 0.59 percent carbon steel (as well as studying the effect of various carbon contents). As is shown in Table 15, martensite exhibited the maximum (a normalized ferrite gave the somewhat higher permeation rate of 0.44 cm³/hr, but it had a low carbon content, of course). In 1957, Amiot reported the results of a thesis study of the effect of microstructure on the permeation of hydrogen through an alloy steel (0.26C, 0.25Ni, 1.02Cr, 0.30Mo). His results have been summarized by Smialowski. (119) The results given in Table 16 show once again a minimum for martensite, but the maximum permeability observed was associated with the globular cementite, or spheroidite.

Thus, it appears that carbide particles in any form hinder the passage of hydrogen through steel, even after steady-state conditions have been reached. Cementite in the coarse spheroidized form appears to have the least effect on permeability; coarse lamellar pearlite offers more resistance to permeation than does fine pearlite; and martensite, in which the carbon atoms are present in the iron lattice but in which there is considerable strain, exhibits the lowest permeability of all.

Effect of Metal Thickness

If the rate of hydrogen flow is determined by the rate of diffusion, then under given conditions. the permeation should be inversely proportional to the thickness of the metal wall. Such a relationship was found by Edwards⁽¹⁶⁴⁾ and by Darken and Smith⁽⁴⁸⁾ who studied the permeation of hydrogen through mild steel during acid pickling. Lewkonja and Baukloh $^{(171)}$ found the same relationship in studying hydrogen flow at high temperatures. According to Smialowski, (119) Raczynski in 1958 studied the permeation of cathodically evolved hydrogen through mild steel membranes of different thicknesses. Raczynski obtained the relation between permeation and the reciprocal of the thickness that is shown in Figure 42. No straightline relationship was observed for sheets less than 1-mm thick. Such deviations from the law of inverse proportionality could be expected for the thinner membranes if surface processes on the entry and exit surfaces were influencing the permeation rate. In addition, using the time-lag method, Raczynski measured the apparent diffusion coefficient (D) of cathodically evolved hydrogen as a function of the thickness of the membrane. These results for mild steel at 194 F (90 C) are shown in Figure 43. For a wall thickness of 0.1 mm, the apparent D-value was of the order of 0.7×10^{-8} cm²/sec. It increased rapidly with increasing thickness until at a thickness of about 0.9 mm, a D-value of about $2 \times 10^{-5} \text{ cm}^2/\text{sec}$ was reached; no further increase was observed for thicker walls. This latter value agrees fairly well with that obtained by extrapolation from high-temperature



FIGURE 42. VARIATION OF THE RATE OF HYDRO-GEN FLOW WITH THE THICKNESS OF A MILD-STEEL MEMBRANE AT 194 F (90 C)(AFTER RACZYNSKI)⁽¹¹⁹⁾



FIGURE 43. EFFECT OF PLATE THICKNESS ON THE VALUES OF D DETERMINED BY TIME-LAG METHOD FOR MILD STEEL CHARGED CATHODICALLY AT 194 F (90 C)(AFTER RACZYNSKI)⁽¹¹⁹⁾

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Structure	Volume of Hydrogen Permeated During 72 Hours, (a) cm ³
Martensite	1.10
Products of isothermal decomposition at 392 F (2	00 C)(b) 1.11
Troostite(c)	2,84
Sorbite(c)	7.70
Normalized steel	8.16
Annealed steel	5,26
Globular cementite	3.95

(a) Electrolyte: 0. 1N H₂SO₄; Temperature: 62. 5-64. 5 F (17-18 C); Cathodic Current Density: 0.014 amp/ cm²; Membrane Thickness: 0.5 mm.

(b) Lower bainite.

(c) A term, no longer used in the United States, the meaning of which is somewhat vague. This term was used formerly to designate various types of structures including intermediate transformation products of austenite decomposition and some tempered martensites.

TABLE 15. HYDROGEN-PERMEATION RATES THROUGH 0.59 PERCENT CARBON STEEL TUBES TREATED TO GIVE DIFFERENT MICROSTRUCTURES(170)

Cathodically charged at room temperature with a cathodic current density of 0.3 amp/cm^2 .

				the second se
Treatment	Structure	Hardness, Vickers	Time to Start of Evolution, minutes	Permeation Rate, cm ³ /hr
Normalized from 1560 F (850 C)	Coarse lamellar pearlite and ferrite	202	39	0.22
Isothermally treated at Very fine pearlite and ferrite 895 F (480 C)		290	35	0.39
Water quenched from 1560 F (850 C)	Martensite	850	45	0.14

TABLE 16. EFFECT OF STRUCTURE ON THE PERMEATION OF CATHODICALLY EVOLVED HYDRO-GEN THROUGH AN ALLOY STEEL (AFTER AMIOT)⁽¹¹⁹⁾

Structure	Permeation Rate,(b) cm ³ /sec/cm ²	
Martensite	3.7 x 10-7	
Tempered martensite	3.8×10^{-7}	
Fine bainite	5.8×10^{-7}	
Coarse bainite	7.2×10^{-7}	
Fine lamellar pearlite	$11 \ge 10^{-7}$	
Coarse lamellar pearlite	8×10^{-7}	
Spheroidite	12×10^{-7}	
	Martensite Tempered martensite Fine bainite Coarse bainite Fine lamellar pearlite Coarse lamellar pearlite	

(a) Austenitized for 30 minutes at 1605 F (875 C) in each case prior to the treatment listed.

(b) Electrolyte: 1N HCl; Temperature: room temperature; Cathodic Current Density: 0.015 amp/cm².

measurements. Smialowski has suggested that the reason that false D-values are obtained when using the time-lag method for thin iron walls is probably that the cathodically evolved hydrogen changes or damages the metal structure in the surface layer and this damaged layer constitutes an appreciable proportion of the total thickness in a thin membrane.

Effect of Cold Work

A number of investigators have studied the effects of elastic and plastic deformation on the rate at which hydrogen permeates steel at room temperature, but only a few have studied their effect on the diffusion of hydrogen. From preceding sections, it is obvious that controversial opinions exist concerning the effect of cold work upon the occlusion of hydrogen in iron and steel. It will be shown in the following discussion that the results obtained for permeability likewise are controversial. Since the majority of investigators have studied permeation of hydrogen in iron and steel by the steady-state permeation-rate method which depends on solubility and diffusivity, it is to be expected that results obtained for diffusivity would be controversial, too.

Chaudron and Moreau⁽¹⁷²⁾ observed that cold work accelerated the permeation of hydrogen during pickling. However, Bastien^(173,174) reported that the permeation of hydrogen is faster for annealed steels than for cold-worked ones during cathodic charging, even though the cold-worked steels absorbed twice as much hydrogen. Rather similar results were obtained by Bhat and Lloyd, (170) who investigated the effects of cold work on the rate at which hydrogen permeated a number of ferritic steels. Although the magnitude of the effects varied from steel to steel, in all cases severe plastic deformation caused a decrease in the permeation rate. Without exception, increasing severity of cold work in the range of 0 to 70 percent reduction resulted in a decrease in permeation rate. A decrease by a factor of 10 as the result of about 70 percent reduction in area was not uncommon. This was accompanied by an increase in the quantity of hydrogen occluded during their cathodic charging. Still another viewpoint was the conclusion arrived at by Darken and Smith⁽⁴⁸⁾ that cold work had no effect on the permeability, though the maximum amount of hydrogen absorbed during pickling and the rate of absorption increased with the amount of cold deformation. When cathodically charging hydrogen through low-carbon steel sheet by the stationary-flow method, Schumann and Erdmann-Jesnitzer⁽¹⁷⁵⁾ observed that the permeation rate first increased with increasing cold work and passed through a maximum at about 10 to 20 percent cold reduction, then decreased rapidly with further deformation; for sheets cold worked 60 to 90 percent, the permeability was nearly zero (see Figure 44). Thus, four markedly different behaviors for the effect of cold work on the permeation of hydrogen through steel have been described.



FIGURE 44. HYDROGEN PERMEATION AS A FUNCTION OF THE DEGREE OF COLD ROLLING⁽¹⁷⁵⁾

Previously, the work of Hudson et al(52, 60, 176) on the effects of cold reduction and heat-treating conditions on the hydrogen solubility (C_s) in mild steel was discussed. These investigators also reported the change in hydrogen permeability, P, produced by these variables. The hydrogensolubility values were taken from the saturation plateaus obtained on the hydrogen-absorption curves they determined. They defined hydrogen permeability, P, as the quantity of hydrogen (cc at standard temperature and pressure) transferred in 1 second through a diaphragm 1 mm thick and 1 cm² in cross section. The values reported were calculated using the equation

$$P = 1.25 \rho \frac{\lambda^2 b^2}{C_s}$$
 (47)

derived by Darken and Smith, (48) but changing the units to fit their experiment. This gave the equation

P = 2.73 x
$$10^{-7} \frac{\lambda^2 b^2}{C_s}$$
, (48)

where λ is the slope of the hydrogen-absorption curve (the hydrogen concentration in the steel plotted against the square root of the time in hours), that is, $cm^3/100 g/(hr)^{\frac{1}{2}}$; C_s is the hydrogen solubility in cm³/100 g; b is the thickness of the specimen in mm; and the density of the steel (p) was taken as 7.8 g/cm³. The effects of cold-reduction and heat-treatment combinations on hydrogen permeability are shown in Table 17 and in Figure 45. Increasing the amount of cold reduction increased the hydrogen solubility for the entire range of reductions used. Increasing the cold reduction also increased the hydrogen permeability until a maximum was reached (below 30 percent reduction for initially hot-rolled steel), after which the permeability decreased as the amount of cold work increased further. These results are rather similar to

Initial Cold Reduction, percent	Box Anneal [4 Hr at 1200 F (650 C)]	Second Cold Reduction, percent	Hydrogen Solubility, cc/100 g	Hydrogen Permeability,(a) P x 10 ⁶
	. 1	¥		
3.7	No	No	4.7	6.3
28.3	No	No	7.1	8.1
47.3	No	No	10.2	4.5
67.6	No	No	20.0	2.6
78.2	No	No	20.9	2.0
2.1	Yes	No	2.1	1.2
28.3	Yes	No	6.2	0.43
47.0	Yes	No	7.3	0.61
67.6	Yes	No	9.0	1.1
78.0	Yes	No	4.8	0.75
3.2	Yes	60	21.2	0.19
28.3	Yes	60	14.2	0.44
45.2	Yes	60	17.9	0.46
66.8	Yes	60	20.2	0.09
78.1	Yes	60	23.0	0.09
78.0	Yes	0	4.8	0.75
78.3	Yes	15	5.2	4.5
78.6	Yes	30	7.7	1.6
78.0	Yes	45	12.8	0,18
78.1	Yes	60	23.0	0.09

TABLE 17. EFFECT OF COLD-WORK AND HEAT-TREATMENT COMBINATIONS ON HYDROGEN SOLU-BILITY AND HYDROGEN PERMEABILITY IN STEEL⁽⁶⁰⁾

(a) Permeability was defined as the quantity of hydrogen (cm³ at standard temperature and pressure) transferred in 1 second through a diaphragm 1 mm thick and 1 sq cm in cross section.



FIGURE 45. EFFECT OF COLD-REDUCTION AND HEAT-TREATMENT COMBINATIONS ON HYDROGEN PERMEABILITY IN STEEL(60)

[HR = hot rolled; CR = cold rolled; BA = box annealed 4 hr at 1200 F (650 C) in a dry atmosphere containing 5 percent hydrogen and 95 percent nitrogen: <u>CR</u> = cold-rolling stage in which various reductions (as plotted in the figure) were used.] those obtained by Schumann and Erdmann-Jesnitzer (mentioned on the preceding page).

De Kazinczy⁽¹⁷⁷⁾ imposed a variable, axial tensile stress on cylindrical tubes that were immersed in an electrolytic cell. He used the steadystate method. The tubes originally were evacuated and the buildup of the pressure of accumulating hydrogen was measured. At stresses below the yield point, there was a reversible change in the permeation rate that was proportional to the applied tensile stress (see Figure 46). The maximum increase in permeation rate due to elastic stress was about 6 percent; it occurred at stresses just below the yield point. An increased rate of flow almost proportional to the applied tensile stress was observed for stresses up to just about 0.2 percent offset yield strength. However, when the specimens were stressed beyond the yield point to at least 0.6 percent plastic elongation, the plastic flow caused the permeation rate to be increased appreciably above the straight-line relationship observed for elastic stresses, as is shown in Figure 47. After the applied stress was removed, specimens that had been permanently strained about 7 percent were found to have acquired a permanent increase in permeability of about 3 to 5 percent. On the

basis of thermodynamic calculations, De Kazinczy concluded that the increased permeation rate resulting from elastic strain was almost entirely due to an increase in solubility in the lattice and not to an increase in diffusivity.* The additional increase in permeation rate accompanying plastic deformation was attributed to stress multiplication in certain zones (probably along slip planes) in which the hydrogen flow was increased.

Elsea and co-workers⁽¹⁴⁸⁾ performed a permeation experiment which indicated that a uniaxial tensile stress of 50,000 psi had no measurable effect on either the diffusion coefficient or the solubility of hydrogen in high-strength SAE 4340 steel. This behavior was shown as follows: an unstressed tubular specimen with a 0.037-inch wall thickness was charged cathodically at a current density of 10 ma/in.² for 24 hours. The current then was charged abruptly to produce a current density of 30 ma/in.² with the specimen remaining unstressed. After a steady-state flow of hydrogen had been obtained, the current was lowered to again produce a current density of 10 ma/in.², and a tensile stress of 50,000 psi was applied. After an extended period, the current again was raised abruptly to produce a current density of 30 ma/in.² The steady-state flow rates of hydrogen at the high and low current densities and the times to obtain a steady-state flow rate of hydrogen following the abrupt change of current were unaffected by the stress, within the limits of experimental accuracy.

Hill and Johnson(14,167) found that the diffusivity in an iron-carbon alloy was decreased manyfold in the temperature range 480 to 750 F (250 to 400 C) by cold working. Since cold working increased the absorption by about the same factor, the permeability should have remained approximately constant. The experiments were performed with iron-carbon alloys that contained 0.16 percent carbon. These investigators found that cold work decreased the diffusivity at temperatures throughout the range from 390 to 750 F (200 to 400 C). Also, the effect increased as the severity of the

*This constitutes further evidence that hydrogen diffuses in annealed steel homogeneously through the lattice.



FIGURE 46. INCREASE OF HYDROGEN FLOW IN ANNEALED STEEL STRESSED ELASTICALLY⁽¹⁷⁷⁾

(Note: q is the amount of hydrogen that passed through a unit length of cylindrical tube per unit time under steady-state conditions.)

cold work increased, just as these investigators found for hydrogen solubility in the same alloy. (14) Part of these results are shown in Figure 32. These workers hypothesized that the effect of cold work on diffusivity would become more severe with increasing carbon content because this trend was found in the solubility measurements. The two short curves in the temperature range 77 to 212 F (25 to 100 C) in Figure 32, when compared with the curve for unstrained pure iron, show the effect of cold work on the diffusivity in pure iron. These curves represent the diffusivity in samples that had been strained plastically 8 and 23 percent in uniaxial tension at room temperature after introduction of hydrogen by pressure charging. Again, as was found for the ironcarbon alloy, the effect of cold work was to reduce the value of the diffusion coefficient. The early stages of strain were proportionately more effective in lowering the coefficient.

These results for cold-worked iron were obtained with samples that had been strained at room temperature. A significant difference was observed when samples were strained at -240 F (-150 C) prior to measurement of the diffusivity. Such deformation had little or no effect on the diffusion coefficient, the values being essentially the same as those obtained from unstrained samples.

In a previous section, work of Darken and Smith(48) was discussed in which they showed that cold work had a pronounced effect on the solubility of hydrogen introduced into steel by acid charging (pickling). A 0.15 percent carbon steel rod was hot worked, being above 1650 F (900 C) in all hotworking operations. This was called "green rod". It was pickled and then baked to remove the hydrogen introduced by the pickling treatment. It was then cold drawn through wire-drawing dies in five steps, with no intermediate anneal. Cold work markedly increased the absorption rate, the time to reach saturation, and the saturation concentration or amount of absorbed hydrogen. Although cold reduction markedly affected absorption, it did not affect the permeability, as calculated from the slope (λ) of the plot of hydrogen concentration



FIGURE 47. INCREASE OF HYDROGEN FLOW IN STEEL SUBJECTED TO PLASTIC DEFORMATION(177)

versus the square root of the time, using the relation

$$P = 9.8\lambda a^2 / (t_s)^{\frac{1}{2}}, \qquad (49)$$

which they derived (a is the radius of the wire and t_s is the time required to reach saturation). This is shown as follows:

Sample Designation	Diameter, inch	Reduction in Area, percent	Permeability, P(a)
Green rod	0.503	0	1.28 x 10-6
After 1st reduction	0.460	16.5	1.26×10^{-6}
After 2nd reduction	0.419	30.6	1.36×10^{-6}
After 3rd reduction	0.369	46.2	1.18 x 10 ⁻⁶
After 4th reduction	0.315	60.8	1.29×10^{-6}
After 5th reduction	0.249	75.9	1.23 x 10 ⁻⁶
		Average	1.26×10^{-6}
Annealed after 5th reduction	0.249	0	<0.24 x 10 ⁻⁶
(a) In N H ₂ SO ₄ .			

Thus, although cold reduction did not alter the permeability, annealing the cold-worked material decreased it by a factor of at least 5.

Some of the most recent work is that of Bryan and Dodge. (49) They studied the permeation rate by maintaining a constant pressure of hydrogen inside a tubular specimen which was placed inside an evacuated jacket. The rate of permeation of hydrogen through a highly purified iron (Ferrovac E) was measured over the pressure range of 1/30 to 300 atmospheres and the temperature range from 259 to 1279 F (126 to 693 C). No difference was observed in the permeability or diffusivity between the tube that had been cold reduced 26 percent and the same tube subsequently annealed. This is in sharp contrast to the case of carbon steels where the diffusivity is sharply decreased by cold working.

Diffusion of Deuterium

Frank, Lee, and Williams⁽⁴⁶⁾ determined the diffusion coefficients of hydrogen and deuterium in SAE 1010 steel for the temperature range from 79 to 187 F (26 to 86 C). A mixture of the two gases was introduced into the steel by abrasion in a mixture of 15 percent normal water (H2O) and 85 percent heavy water (D2O); this composition was selected because it provided nearly equal permeation rates for hydrogen and deuterium. A mass spectrometer was used to measure simultaneously the rates of evolution of H2, HD, and D2 from the steel. The diffusion coefficients were measured by observing the rates of decay of the diffusion rates as a function of elapsed time after the process of introducing the isotopes was stopped. A plot of the diffusion coefficients for hydrogen and deuterium in the steel is shown in Figure 48. The ratio of the diffusion coefficient of hydrogen to that of deuterium was found to be constant within the limits of experimental error over the entire temperature range investigated; it had a value of 1.37 ± 0.02 ,

which is very close to the square root of the ratio of the masses (1.414) which one might expect to get on the basis of simple kinetic theory. This ratio held, even at the lower temperatures where the curves in Figure 48 showed some deviation from a straight line. However, the ratio of the <u>permeation</u> rates varied with variation in temperature (see the tabulation below), reflecting the variation in the ratio of the solubilities of hydrogen and deuterium with temperature.

Temp	erature	
C	F	$\underline{P}_{H}/\underline{P}_{D}$
86	187	9.5
80	176	10.7
79	174	10.1
60	140	6.1
60	140	6.8
52	126	4.4
52	126	4.6
50	122	6.7
42	108	5.8
34	93	4.9
26	79	4.5
26	79	4.8

Within the limits of experimental error, there was no difference in the activation energies for hydrogen and deuterium diffusion (the two curves in Figure 48 having the same slope). In a previous paper, Frank and co-workers pointed out that the evolution of hydrogen from steel may not result from simple interstitial diffusion. If it does not, then one need not expect to get a difference in activation energies for hydrogen and deuterium as other authors have suggested. The results of these experiments indicated that, whatever the mechanism of the diffusion process at and near room temperature, the principal difference in the diffusion coefficients of hydrogen and deuterium is the result of the mass effect, and this does not vary with temperature.



FIGURE 48. CURVES OF 1nD PLOTTED AGAINST 1/T x 1000 FOR HYDROGEN AND DEUTERIUM(46)

Apparently, no data for the diffusivity of the third isotope, tritium, in steel are generally available.

Removal of Hydrogen

Although under certain corrosive conditions the hydrogen content of steel will increase during storage, in normally dry air hydrogen is constantly being lost from a piece of ferritic or martensitic steel until the hydrogen content of the steel reaches equilibrium with its environment. The rate of loss depends on, and varies greatly with, such factors as temperature, thickness of the material, whether the steel is ferritic or austenitic, and the amount of cold work.

The Effects of Temperature, Section Size, and Steel Composition

The loss of hydrogen from an austenitic steel is imperceptible at room temperature. Carter(178) found that no measurable quantity of hydrogen escaped at room temperature from austenitic samples in a period of 3 months, even though they contained 2 to 5 times as much hydrogen as the ferritic samples from which as much as 61 percent of the total hydrogen content was evolved in 10 days.

In general, the rate of hydrogen evolution from all steels increases with increasing temperature, at least within a few hundred degrees of room temperature. Hydrogen has been observed escaping from 5/8-inch-diameter cylinders of chill-cast ferritic steel at room temperature within 10 to 15 minutes after casting. (179) In 2 weeks at room temperature, such specimens lost as little as 10 percent or as much as 80 percent of their total hydrogen. The rest of the hydrogen evolved extremely slowly, if at all. (180,181) From these latter investigations, it appears that aging to remove hydrogen from ferritic steel is roughly 250 to 500 times faster at 400 F than it is at room temperature, and that the aging time increases as the 3/2 power of specimen size. This means that the time increases directly with increasing volume and inversely with increase in surface area. Therefore, for a cylindrical bar, the time increases as the square of the diameter. Hence, removal from large masses is very slow. For 1-inch-square cast coupons of carbon steel, 60 days at room temperature was required to reduce the hydrogen to the "equilibrium" level, whereas it was estimated that 6 years would be required for 4-inch coupons. (180, 181, 182) At 400 F, the time required to accomplish the same results were 6 hours and 100 hours, respectively.

Hobson(160,183) has studied the removal of hydrogen from steel, with particular attention to its removal from large masses. The marked effects of temperature and the effects of holding time at a given temperature on the residual fraction (U) of hydrogen remaining in the steel are shown in Figure 49. Note that two different time scales have been used to show the extremes of rate. Based on the time to reach U = 0.5, and taking the rate at room temperature as unity, the rate at 392 F (200 C) was 209 times faster and the rate at -110 F (-78 C) was 1/50 as fast (as is shown in Table 18). From these results, it is obvious that the theoretical equation (of Sykes) is extremely misleading at temperatures below 392 F (200 C)*. The effects of section size (cylinder diameter) are shown in Figures 50 and 51 for two different steels. In Figures 50a and 51a, the residual fraction of hydrogen has been plotted directly against time. In Figures 50b and 51b, the residual fraction has been plotted directly against time reduced to a common diameter of 11.25 mm. Theory predicts that for the latter arrangement the curves for a given steel should coincide. In practice, there was considerable scatter of the evolution curves, but this scatter was random for the different sizes of steel. It was concluded that the square law is followed approximately. Comparison of the two steels of Figures 50 and 51 shows that for a specimen of a given size the evolution was slower by a factor of about 40 for the 14Cr steel than for the 2. 5Ni-Cr-Mo steel. Table 19 compares the results for all 9 steels studied.

*See the section dealing with the effect of temperature on diffusion of hydrogen in steel in which it is shown that anomalously low diffusivities are encountered at temperatures below 750 F (400 C).





FIGURE 50. (a) EFFECT OF CYLINDER DIAMETER ON EVOLUTION RATE FOR 2.5%Ni-Cr-Mo STEEL (b) EVOLUTION CURVES REDUCED BY INVERSE SQUARE LAW TO COMMON TIME SCALE⁽¹⁶⁰⁾



FIGURE 51. (a) EFFECT OF CYLINDER DIAMETER ON EVOLUTION RATE FOR 14%Cr STEEL (b) EVOLUTION CURVES REDUCED BY INVERSE SQUARE LAW TO COMMON TIME SCALE⁽¹⁶⁰⁾

TABLE 18. EFFECT OF TEMPERATURE ON THE RATE OF HYDROGEN DIFFUSION IN 2.5%Ni-Cr-Mo STEEL $A^{\rm (160)}$

Tempe	erature	Time to Reach	Rate Relative to RT	Relative Rate Predicted	Diffusion
С	F	U=0.5, hr ^(a)	Taken as Unity	by Sykes' Equation	Constant
-78	-172	~1250	0.018	0.14	5.8 x 10 ⁻⁹
RT	RT	23	1	1	3.2×10^{-7}
50	122	5.5	4.2		1.35 x 10-6
100	212	0.55	41.9	2.3	1.34×10^{-5}
200	392	0.11	209	4.5	6.7×10^{-5}

(a) For 11.25-mm cylinder; 45-mm results are divided by 16 to be comparable.

CALCULATED TIMES FOR CYLINDERS OF 11.25 MM DIAMETER OF STEELS A-I TO
REACH VARIOUS VALUES OF RESIDUAL HYDROGEN FRACTION U(160)

		Number of	the second se	ime to = 0.7		and the second se	= 0.5		and the second se	= 0.3	
Steel	Туре	Experiments	Mean	Max	Min	Mean	Max	Min	Mean	Max	Min
A	2.5% Ni-Cr-Mo	10	7	10	5	19	24	14	44	58	33
в	14% Cr	7	690	1320	310	2080	2960	1500	5640	6720	4560
С	2.5% Ni-Cr-Mo	1	11			28			78		
D	0.34% C	1	32			63			108		
E	0.8% C-1.0% Cr	2	11	12	10	27	31	22	72	88	56
F	0.8% C-Cr-Mo	1	26			62			150		
G	0.8% C-Cr-Mo	1	20			35			85		
н	3% Cr-Mo	2	48	60	35	108	140	76	250	360	145
I	3% Cr-Mo	1	85								

In the section of the present report on solubility, it was shown that the solubility of hydrogen in alpha iron increases rapidly with increasing temperature. Therefore, baking or aging treatments to remove hydrogen from ferritic steel must be a compromise between greater permeability or rate of diffusion obtained at increased temperatures, so as to shorten the treatment time, and the lower solubility prevailing at lower temperatures. For parts that are not too thick, a baking treatment of 24 hours at about 375 F frequently is used. Retained austenite has a high equilibrium solubility for hydrogen, so hydrogen tends to concentrate in this constituent. This hydrogen cannot be removed by baking unless the treatment also transforms the retained austenite.

Moore and Smith(184) have presented a review of the early work on occlusion and evolution of hydrogen in and from iron and steel. They also studied the evolution of hydrogen from carbonyl iron strips cold rolled to a thickness of 0.06 mm. When this material was exposed to pure hydrogen at 1 atmosphere and 77 F (25 C) for 55 days, it was found that 200 times as much hydrogen was occluded as could be accounted for by true lattice solubility. The hydrogen then was evolved from the foil in an evacuated container by holding first at room temperature and then by heating to successively higher temperature levels and holding isothermally at each temperature until the evolution slowed down perceptibly. The results of such a test are shown in Figure 52. They found that at each temperature level there was rapid evolution at first, but it then slowed down and practically stopped. When the temperature was raised, the pattern was repeated, except that more gas was evolved with each successive temperature level up to 1965 F (1075 C). This behavior is contrary to what might be expected in view of the fact that solid solubility increases with increasing temperature. However, heavily cold-worked iron or steel is capable of occluding very large contents of hydrogen, above the true solubility limit, and these investigators suggested that these data indicated that

it is held occluded in rifts rather than in lattice solution.



FIGURE 52. ISOTHERMAL EVOLUTION FROM PURE IRON FOIL⁽¹⁸⁴⁾

Geller and Sun⁽³⁸⁾ investigated the effusion of hydrogen at room temperature. They found no evolution at all from austenitic steels, and they concluded that practically all alloy additions to ferrite retard the effusion of hydrogen and that structure also plays a big part. The effect of increasing amounts of silicon and of a high chromium content in retarding the effusion of hydrogen from ferrite is shown in Figure 53.

The effects of temperature on hydrogen removal also are indicated by the results of an experiment performed by Zapffe and Sims. (185) A group of 0.505-inch-diameter, mild-steel tensile specimens was heated in hydrogen for 3 hours at 2010 F (1100 C) and then guenched in water to retain a maximum amount of hydrogen. These charged specimens were then treated for varying periods of time at temperatures from 212 to 1470 F (100 to 800 C) by 180 F (100 C) increments. At the end of the treatment, each specimen was again quenched so that hydrogen loss on cooling would not affect the results. The specimens were immediately tested in tension. The embrittlement, as measured by loss in reduction in area, disappeared faster as the temperature was raised. Ductilities of the hydrogen-free control group heat treated at the same temperatures served as endpoint ductilities for the hydrogen-removal studies. The curve in Figure 54 shows the variation with temperature of the length of anneal necessary to obtain relief from hydrogen embrittlement, that is, for obtaining near-maximum ductility.

The early investigations of hydrogen embrittlement induced by pickling in acid solutions or by electrolysis also showed that aging at room temperature or heating to moderately elevated temperatures (known as baking) causes a gradual recovery of the original ductility. Some investigators reported that complete recovery had been obtained, while others found that recovery was incomplete at either room temperature or at elevated temperatures. However, this early work was done largely with low-carbon steel.

The method of introducing hydrogen into the steel has been found to have little effect on the rate of recovery at room temperature. For example, it was found to be only slightly different for lowcarbon steel wire embrittled by cathodic treatment from that for wire embrittled by heating in hydrogen. (186) However, variation in the crosssectional area of the test piece has a large effect on the rate of recovery at either room temperature or a given elevated temperature because the recovery depends on the diffusion of hydrogen. For this reason, elevated temperatures are increasingly more effective. A severely embrittled low-carbon steel wire 0.040 inch in diameter was found to recover 50 percent of the initial bending capacity in about 1 week, while a wire of the same composition



AT ROOM TEMPERATURE(38)

but 0.160 inch in diameter recovered to only 40 percent of the original bend value in 3 months. (186)

Apparently, low-alloy steels, heat treated to high strength levels, completely recover the ductility lost by hydrogen introduced by acid pickling if they are stored for a long time before use, or if they are baked at a sufficiently high temperature, provided that they are not electroplated after pickling. For example, tensile tests of SAE 4340 steel, heat treated to a hardness of Rockwell C 47, indicated that complete recovery from the effects of pickling in HCl for 1 hour was accomplished by aging at room temperature for 5 hours or more (see Figure 55).



FIGURE 54. RELATION OF TEMPERATURE TO ANNEALING TIME NECESSARY FOR REMOVING HYDROGEN EMBRITTLE-MENT FROM MILD-STEEL SPECI-MENS(185)

(Tensile specimens 0.505 inch in diameter.)



FIGURE 55. EFFECT OF AGING ON THE HYDRO-GEN EMBRITTLEMENT OF PICKLED AND CHROMIUM-PLATED TENSION SPECIMENS OF 4340 STEEL (0.505-IN. DIAM.)(DATA BY NORTH AMERICAN AVIATION)(187)

Effects of Environment and Surface Reactions

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One way of looking at diffusivity, D, is that it is a measure of the time required for a specimen of given size under given conditions to absorb a given fraction of the hydrogen it would absorb in a very long time. In the case of constant boundary conditions and zero initial hydrogen content in the steel, the solution of Fick's diffusion equation (for specimens of similar shape) shows that the hydro gen concentration at any given point is a function Dt/a^2 , where t is time and a is a linear dimension of the specimen (for example, the radius of a sphere or of a long cylinder, or the thickness of a slab or sheet). From this relation, it is clear that, for constant diffusivity, the time required for a specimen to absorb a given fraction, say half, of the total amount of hydrogen that it would absorb in a very long time (independent of that amount) would be proportional to the square of the dimension. For example, one would expect that it would take 4 times as long to half-saturate a 1-inch rod than it would to half-saturate a $\frac{1}{2}$ -inch rod. Likewise, for constant diffusivity, the reverse process (removal of hydrogen from the saturated specimen) would be expected to be similar, and the time required for the removal of any given percentage of the saturation amount of hydrogen should be the same as that required for the absorption of the same percentage by the same specimen at the same temperature. However, surface reactions, rather than diffusion through the bulk of the metal, may be controlling the entry and removal of hydrogen; the reactions may be different in the two cases. A number of studies have shown that environment is important in the removal of hydrogen from steel.

Bastien(188) studied the influence of the environment on the recovery of the bend characteristics of steel specimens embrittled by pickling in HC1. Figure 56 shows the markedly faster rate of recovery achieved when the specimens were immersed in water or were in air saturated with water vapor, than when they were in dry air. Immersing the specimens in water to which 0.01 percent of Na₂S was added retarded recovery the most. In this latter instance, corrosion may have occurred to provide an opposing reaction (hydrogen entry). The addition of Na₂S poison to the pickling acid increased the brittleness of the steel and it also increased the time required for complete recovery after a given pickling time.

Evans⁽⁶⁴⁾ stated that immersion in water is more effective in removing hydrogen from pickled steel than is exposure to hot air. Of course, both are considerably more effective than storage in air at room temperature. Slater⁽¹⁸⁹⁾ pointed out the marked superiority of an environment of water at 212 F (100 C) over air at the same temperature as a means of eliminating hydrogen. Other reports of a qualitative nature have indicated that hydrogen removal is more rapid in water than in air. Morris⁽¹²⁸⁾ showed that an oxide film on a steel surface not in contact with acid solution resists the passage of hydrogen generated on the other surface by reaction with acid. The rate of recovery of chromium- and cadmium-plated stainless steel (17Cr-1C) wire from the embrittled condition was shown by Zapffe and Haslem(132) to be more rapid at 212 F (100 C) for specimens stored in water or 3 percent NaOH solution than for those stored in oil or dry argon.

Darken and Smith(48) studied hydrogen absorption during acid pickling and evolution of hydrogen on removing the steel from the acid and holding at the same temperature used for charging. The evolution rate of absorbed hydrogen was slower than the rate of absorption for several experiments which involved at least three different charging conditions. In one instance, 5 days' aging time was required for loss of half the hydrogen at 95 F (35 C), whereas only 7 hours' charging was required for the absorption of the same amount of hydrogen. Another comparison is shown in Figure 57. The difference between water-treated and air-treated specimens was not large, which was attributed to the fact that considerable time was required to heat the specimens up to 212 F (100 C) in air.

Darken and Smith explained the observed lower rate of evolution of hydrogen as compared with the rate of absorption on this basis. They assumed that the permeability is proportional to the gradient of the lattice-dissolved hydrogen and not the total hydrogen content. In these acidcharged specimens, the total hydrogen content was much greater than the portion dissolved in the lattice. As the amount of hydrogen in the steel decreased by effusion, the amount of lattice-dissolved hydrogen (the gradient of which is the driving force) decreased much more rapidly, and thus escape was very slow. They hypothesized that the observed variation does not correspond to a change of the diffusivity of lattice-dissolved hydrogen atoms in a perfect iron lattice, but, rather, that the change is caused by a "trapping" of a fraction of the hydrogen



FIGURE 56. INFLUENCE OF THE NATURE OF THE EXTERNAL MEDIUM ON THE RECOVERY OF THE BEND CHARACTERISTICS AFTER PICKLING⁽¹⁸⁸⁾

atoms in imperfections of the iron lattice. They pointed out that it might be thought that the observed phenomena could be accounted for by the supposition that gaseous (molecular) hydrogen at high pressure accumulates in macroscopic voids. If this were so, they reasoned that the "amount" so accumulated would be proportional to the square of the concentration of lattice-dissolved hydrogen (Sieverts' law), and, hence, the "amount" in the voids would be a negligibly small fraction of the total hydrogen at low concentrations and the diffusivity would be greater than at higher concentrations.* This is contrary to observed behavior which indicates definitely that diffusivity is greater at higher concentrations than at lower. This conclusion of theirs that a large part of the hydrogen is somehow trapped is strengthened by the observed large increase of absorbed hydrogen resulting from prior cold work.

Hudson, Norris, and Stragend⁽⁸¹⁾ studied the effect of environment on the rate of removal of hydrogen from a commercial, low-carbon steel. Their work involved specimens precharged with hydrogen only by acid pickling in 2N H₂SO₄ at 100 F (38 C) for 24 hours, a time found to be sufficient to ensure saturation with hydrogen. Postcharging treatments included treatment in deoxygenated distilled water at 39, 100, 147, and 201 F (4, 38, 64, and 94 C), distilled water saturated with oxygen used at 100 F, and aqueous solutions of the following oxidizing agents used at 100 F:

*The present authors do not fully agree with this reasoning. Sievert's law expresses an equilibrium relationship between the concentration of atomic hydrogen in the lattice and the pressure of molecular hydrogen external to the iron lattice. This relation usually is expressed in the form: the concentration or solubility is proportional to the square root of the pressure of molecular hydrogen, but it may also be expressed in the form: the pressure is proportional to the square of the concentration or solubility (see Equation 1, page 5). Since Darken and Smith made no mention of the size and number of the voids other than that the voids are macroscopic in size, there is no way of knowing what the "amount" of hydrogen in the voids will be. Therefore, the relationship between the "amount" in the voids and the amount in the lattice also is unknown. Where the volume of the voids is a very small fraction of the volume of the iron (as might be expected to be the case in wrought products of high quality), Darken and Smith's statement about amount probably is correct, since a low pressure in a very small volume would correspond to a small "amount" of hydrogen gas. When one considers that the amount in the lattice is measured in terms of a few ppm by weight, it is apparent that the size and number of the voids for which the statement will apply is strictly limited. The equilibrium statement does imply that at low concentrations the pressure in the voids will be very low.

Solution	<u>M</u>	Solution	<u>M</u>
KMnO ₄	0.01	K ₂ S ₂ O ₈	0.1
KMnO ₄	0.1	NaNO ₂	0.1
KMnO ₄	0.25	HNO ₃	0.1
K2Cr2O7	0.1	HNO ₃	15.6
KCrO ₄	0.1	(NH4)2Ce(NO3)6	0.1
KClO3	0.1	Fe2(SO4)3(NH4)2SO4	0.1
KBrO3	0.1	K ₃ Fe(CN) ₆	0.1
KIO ₃	0.1	H ₂ O ₂	3.0
KCIO4	0.1	NaAsO2	0.1
KIO ₄	0.1	NaHAsO4	0.1
NaBO3	0.1	1	

Other environments studied included water-dioxane mixtures (containing 2, 10, 20, 40, 60, and 80 percent water), dioxane, methanol, dry nitrogen, humidified nitrogen (0.63 and 1.86 percent water), dried hydrogen gas, and a vacuum. For all of these environments, removal-rate studies were performed at 100 F. Additional temperatures from 32 to 212 F (0 to 100 C) were investigated for some of the environments so as to permit estimates to be made of activation energies. Following all treatments, residual hydrogen was determined by the warm-extraction technique.

The effects of postcharging treatments with aqueous solutions at 100 F on the rate of removal of hydrogen from steelare summarized in Table 20. The rate of removal was highest when the steel was immersed in water. These investigators stated that this observed behavior agrees with the findings of Lewis and Ubbelohde. (190) For aqueous solutions of oxidizing agents, no treatment was better than deoxygenated water for hydrogen removal. These



FIGURE 57. COMPARISON OF RATE OF ABSORP-TION (CURVE A) AT 95 F (35 C) WITH RATE OF EVOLUTION (CURVE B) AT 95 F INTO MERCURY AND (CURVE C) INTO AIR AT 212 F (100 C) AND (CURVE D) WATER BATH AT 212 F⁽⁴⁸⁾

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TABLE 20. EFFECT OF THE ENVIRONMENT (CONSISTING OF AQUEOUS SOLU-TIONS OF VARIOUS OXIDIZING AGENTS AND DEOXYGENATED WATER) ON THE RATE OF REMOVAL OF HYDROGEN FROM LOW-CARBON STEEL(81)

		Hydrogen Removed, percent		
Treating Solution	M	2 hr	4 hr	
K ₂ CrO ₄	0.1	15	14	
K2Cr2O7	0.1	7	14	
KBrO3	0.1	17	15	
H ₂ O ₂	3.0	17	18	
KMnO ₄	0.25	18	22	
KIO3	0.1	15	23	
NaBO3	0.1	20	24	
NaAsOz	0.1	21	24	
$Fe_2(SO_4)_3(NH_4)_2SO_4$	0.1	28	25	
KMnO ₄	0.1	16	27	
NaNOZ	0.1	18	30	
K3Fe(CN)6	0.1	20	32	
K ₂ S ₂ O ₈	0.1	49	42	
KIO4	0.1	18	44	
KC1Ô3	0.1	34	62	
KMnO ₄	0.01	58	66	
$(NH_4)_2 Ce(NO_3)_6$	0.1	59	74	
Na ₂ HAsO ₄	0.1	56	74	
KCIO4	0.1	65	81	
HNO ₃	15.6	64	84	
HNO ₃	0.1	49	85	
H ₂ O (O ₂ -saturated)		68	80	
H ₂ O (deoxygenated)		72	85	

data indicate that the oxide films which often form in visible amounts on the steel surface during treatment hinder the removal of hydrogen from steel. The linear relationship obtained for hydrogen removal in a water environment when 1/c(where c = concentration of hydrogen in steel) was plotted against time shows that a second-order rate law applies for this treatment (see Figure 58), that is, $dc/dt = -k_2c^2$. Note how much more rapid the removal rate became at slightly elevated temperatures. The rate constant more than doubled for a 36 F (20 C) rise in temperature. Removal of hydrogen in water-dioxane mixtures depended directly on the amount of water present in the medium.

The marked effect of environmental conditions on the rate of hydrogen removal from steel is well demonstrated by a comparison of the rates from steel immersed in water and steel held in a vacuum, both at 100 F, as shown in Figure 59. The behavior in the two environments indicated that the hydrogen removal responded to different rate laws. The difference between the initial slopes was significant, that for vacuum being very small. The time required to remove 50 percent of the





(The linear relationship shows that a secondorder rate law applies for hydrogen removal in a water environment; c = concentration of hydrogen in steel.)



FIGURE 59. PLOT SHOWING THAT THE REMOVAL OF HYDROGEN FROM STEEL IN WATER AND IN A VACUUM OBEYS DIFFERENT RATE LAWS⁽⁸¹⁾

(The difference between initial slopes is significant; c = concentration of hydrogen in steel.) hydrogen from the steel investigated was only l hour for specimens immersed in water and 5.8 hours for specimens held in a vacuum at the same temperature.

The fact that such large differences in hydrogen removal rate do exist shows that surface reactions can be rate determining. If the process were simple and diffusion controlled, the amount of moisture present on the steel surface could not affect the rate because it cannot affect the rate of hydrogen diffusion within the iron lattice.

For environments containing appreciable moisture, a second-order rate law was found to apply for the process of hydrogen removal from steel, as was discussed above. However, for environments containing extremely small amounts of moisture, a first-order rate law applied, as indicated by linear plots obtained when ln (l/c) was plotted versus t. Such plots were obtained for dried hydrogen and dried nitrogen at several different temperatures, for storage in a vacuum at 100 F, and for treatment in dioxane and methanol at various temperatures. A plot showing part of these results is given in Figure 60 (removal of hydrogen from steel in water, a second-order reaction, is shown for comparison). Note that hydrogen was readily removed from steel in a dryhydrogen atmosphere. This is possible because hydrogen atoms can diffuse against a high pressure of molecular hydrogen, if the partial pressure of dissociated hydrogen is low, as it is at nearambient temperatures.



FIGURE 60. SEMILOGARITHMIC PLOTS SHOWING THAT A FIRST-ORDER RATE LAW (LINEAR RELATIONSHIP) APPLIES FOR HYDROGEN REMOVAL FROM STEEL AT 100 F (38 C) IN ENVIRON-MENTS LOW IN MOISTURE⁽⁸¹⁾

(Removal of hydrogen from steel immersed in water, a second-order reaction, is shown for comparison.) /229 Although it has not been proved that the rate of hydrogen removal from steel in dry environments is diffusion controlled, for the low-carbon steel used in this investigation apparent diffusion coefficients calculated from measured first-order rate constants were in the range determined experimentally by other investigators for rather similar materials at this same temperature.

In another paper, which was part of the study of hydrogen absorption and desorption just referred to, Hudson et al(52) confirmed the importance of steel surface reactions in the desorption of hydrogen from steel. The rate of hydrogen desorption as a function of cold work and environment is shown in Figures 61 and 62 (note the change in the rate scale), and in Table 21. A maximum value of hydrogen desorption rate in water and in dry nitrogen at 100 F was observed at 20 to 30 percent cold reduction. Severely cold-worked steel exhibited a considerably lower desorption rate than less severely cold-worked or annealed materials (the effect of annealing is shown in Table 22), regardless of whether the environment was water or dried nitrogen. For reductions less than 67 percent, the desorption was much faster in water than in dried nitrogen. Similar effects of cold reduction and environment on rate of hydrogen desorption were obtained by these investigators for SAE 1020 killed and capped steels in still another investigation, (176) Although the killed steel had a higher saturation solubility for hydrogen on pickling, for most conditions it exhibited the lower rate of desorption.

Owe Berg⁽¹⁹¹⁾ has suggested that migration of hydrogen atoms in steel is sufficiently rapid to maintain a nearly uniform hydrogen distribution in the metal during absorption and desorption, and that diffusion in the metal is very rapid compared with surface processes. The reader is referred to Barton's paper⁽¹¹⁷⁾ on the mechanism of transport of hydrogen across a solution-metal interface (previously discussed in the present report under cathodic charging in the section on hydrogen entry) for a discussion of rate-controlling mechanisms.

Hudson and co-workers pointed out that if Owe Berg's hypothesis is valid, neither the firstorder rate law that they observed for dry environments nor the second-order rate law observed for moist environments could result from diffusion control, and in both cases, then, surface reactions presumably would be involved. In all estimates of diffusivity, the assumption is made that surface reactions may be neglected. Yet this work shows that in the presence of moisture this is not always the case. Therefore, the investigators suggested that some of the differences in reported diffusionrate measurements may result from surface effects.

Differences in steel composition, surface roughness, surface area, and geometry also influence the rate of removal of hydrogen from steel, and Hudson et al were planning to investigate these factors in a continuation of their work.





FIGURE 61. CHANGE IN REMOVAL RATE OF HYDROGEN FROM STEEL IMMER-SED IN WATER AT 100 F (38 C) AS A FUNCTION OF COLD WORK⁽⁵²⁾

FIGURE 62. CHANGE IN HYDROGEN REMOVAL RATE FOR STEEL IN AN ENVIRON-MENT OF DRY NITROGEN AT 100 F (38 C) AS A FUNCTION OF COLD WORK⁽⁵²⁾

RATE OF HYDROGEN DESORPTION FROM STEEL AS A FUNCTION
OF COLD WORK ⁽⁵²⁾

	After 30-Minute Pickle in 2N H ₂ SO ₄ at 100 F (38 C)		Desorption Rate ^(a)	
Reduction, %	Hydrogen Concentration, cc/100 g	Weight Loss, %	Water at 100 F (38 C), $k_2 \times 10^3$	Dry N ₂ at 100 F (38 C), $k_1 \times 10^5$
3.7	4.7	2.5	27.3	4.7
28.3	7.1	2.4	36.8	6.1
47.3	10.2	2.1	20.6	5.4
67.6	20.0	2.3	4.1	4.1
78.2	20.9	2.2	1.4	2.4

(a) Unit for k2 is g/cc(STP)sec; unit for k1 is reciprocal seconds.

TABLE 22. RATE OF HYDROGEN DESORPTION FROM STEEL C AS A FUNCTION OF PROCESSING CONDITION⁽⁵²⁾

	Time Pickled	Hydrogen		Desorption Rate(a)	
Condition	in 2N H ₂ SO ₄ at 100 F (38 C), min	Concentration After Pickling, cc/100 g	Weight Loss, %	Water at 100 F (38 C), k ₂ x 10 ³	Dry N ₂ at 100 F (38 C), $k_1 \ge 10^5$
Cold reduced	8	19	0.22	6.4	2.2
Continuously annealed	15	12	0.23	18.2	5.2
Box annealed	20	5	0.21	55.8	10.7

(a) Unit for k_2 is g/cc(STP)sec; unit for k_1 is reciprocal seconds.

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Effects of Coatings

In addition to the effects that occur when hydrogen is concurrently plated with a desired metal electroplate (marked reduction in ductility and development of susceptibility to delayed hydrogen cracking), the plate has another effect. The presence of a more-or-less impermeable metal coating, such as cadmium, makes the evolution of hydrogen from the interior of the base metal more difficult; this may serve to aggravate the effects of embrittlement and delayed failure of electroplated steel. Although an appropriate baking treatment may restore most or all of the ductility to the plated steel, often such a treatment does not overcome the propensity toward delayed brittle failure.

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R. D. Johnson et al(140) showed that the delayed-failure behavior was almost identical for commercial cadmium-plated steel and the same steel cathodically charged with hydrogen, for the conditions used. However, the room-temperature aging characteristics of cadmium-plated specimens and hydrogen-charged nonplated specimens were markedly different; ductility was recovered more slowly in the case of the cadmium-plated specimens. Figure $55^{(187)}$ shows that chromium plate serves as a barrier to hinder evolution of hydrogen on aging, and, hence, to hinder recovery of ductility. Figure $63^{(187)}$ which shows recovery by baking at 375 F demonstrates how the chromium plate also hinders recovery under these conditions, and further, that the cadmium plate commonly used for corrosion protection of high-strength steel parts acts as an even more effective barrier. The following data obtained by H. H. Johnson et al(192) also show the barrier effect of cadmium plate:

Recovery Characteristics of Plated and Deplated Specimens

(Normal, unplated ductility = 40-42.5 percent reduction in area.)

	Reduction in Area, percent		
Baking Treatment	Deplated Chemically	As Plated	
300 F, 0.5 hr	41.5	10.5	
300 F, 2 hr	42.5	13.5	

Elsea and co-workers(118) performed experiments to determine the effect of a cadmium plate on the absorption and on the discharge of hydrogen by steel. A cadmium plate approximately 0.0003 inch thick was electrodeposited on one side of a permeability specimen. Both the cyanide and the fluoborate plating processes were used in plating the specimens. Several of the specimens were tested by using the cadmium-plated side of the specimen as the entrance surface or cell side. The remaining specimens were tested with the cadmium plate on the exit surface. Without exception, when



FIGURE 63. EFFECT OF BAKING TIME ON HYDROGEN EMBRITTLEMENT OF CHROMIUM-, CADMIUM-, AND ZINC-PLATED SPECIMENS (DATA BY NORTH AMERICAN AVIATION)⁽¹⁸⁷⁾

(4340 Steel, Hardness = Rockwell C 45.)

the cadmium plate was on the entrance surface, there was no absorption of hydrogen, even after several hours of charging with current densities of 10 to several hundred ma/in.² for the poisoned 4 percent sulfuric acid electrolyte and 500 ma/in.² for the sodium hydroxide electrolyte. However, when the specimens were placed so that the cadmium was on the exit surface, the flow of hydrogen through the specimen was restricted until the internal pressure of hydrogen blistered the cadmium plate. Hydrogen then permeated the specimen as though there were no plate present. The behavior was the same whether the cadmium plate was applied by the cyanide or the fluoborate process.

Morris, ⁽¹²⁸⁾ in studying the permeation of hydrogen into steel by corrosion in 1 percent citric acid solution, discovered that a coating of tin on the side of the sheet not in contact with the acid (the exit side) retarded permeation of hydrogen through the steel. Effusion was not stopped completely, probably owing to discontinuities in the tin layer, he suggested. A film of oxide also retarded effusion, but a coating of lacquer had little effect.

In studying the rates of absorption and effusion of hydrogen obtained for α -iron, Carmichael et al⁽¹⁹³⁾ found that thin oxide films lowered the apparent diffusivity of annealed iron markedly.

Other investigators⁽¹⁹⁴⁾ determined the hydrogen permeability of Types 321 and 430 stainless steel and Hastelloy B, calorized Type 430 stainless steel, glass-coated Type 304 stainless

steel, and three Hastelloy B samples coated with a proprietary material. The tests were performed at temperatures in the range of 1020 to 1510 F (550 to 820 C) and with hydrogen pressures of 1.1, 1.5, and 2.0 atm. The permeabilities of the uncoated specimens were uniformly much greater than those of the surface-treated metals. A 3 to 7-mil coating of a suitable, high-temperature glass enamel or 10 mils of surface calorizing was 10 to 100 times more effective as a hydrogen barrier than was 60 to 120 mils of untreated steel.

The recovery of the original properties of cadmium-plated high-strength steel parts is an important problem in the aircraft industry. At room temperature, the recovery of ductility that was lost as the result of hydrogen embrittlement incurred during cadmium plating has been found to be extremely slow and often is incomplete after very long aging times. For example, cadmiumplated rings cut from a tubular part made of SAE 4340 steel with a hardness of Rockwell C 50 showed no recovery of ductility after being aged for 2 weeks at room temperature (see Figure 64). However, the recovery of ductility in embrittled. cadmium-plated steel parts increases rapidly with increasing temperature, as Eakin and Lownie(129) showed for thin, cadmium-plated clock-spring steel. This is illustrated in Figure 65. They also found that, as the hardness level of the steel increased, the embrittlement became more severe and higher aging temperatures were required to produce complete recovery in a bend test. In a study of delayed failures in zinc-plated lockwashers, Valentine⁽¹³⁰⁾ found that a subsequent aging treatment at 400 F eliminated or reduced the delayed failures. However, in an investigation of many thousands of acid-descaled and cadmium-plated steel dome lockwashers, Stefanides(131) found no evidence that baking after plating was very helpful. Zapffe and Haslem⁽¹³²⁻¹³⁴⁾ embrittled mild steel and hardenable stainless steel (17 percent chromium, 1 percent carbon) wires by electroplating with cadmium, chromium, zinc, tin, nickel, or lead. They found that baking treatments were effective in recovering the original ductility. Figelman and Shreider(136) studied the effects of electroplating with chromium, copper, nickel, zinc, or cadmium on the ductility of heat-treated spring steel. The testing method consisted of measuring the bend angle necessary to produce failure of a flat specimen. They found that aging at 390 to 570 F (200 to 300 C) was almost always effective in recovering the ductility lost as a result of the introduction of hydrogen into the base steel.

Although investigations such as the one on clock-spring steel by Eakin and Lownie indicate that temperatures up to at least 525 F greatly accelerate recovery of ductility lost as the result of



FIGURE 64. EFFECT OF ROOM-TEMPERATURE AGING AND BAKING AT 375 F ON HYDROGEN EM-BRITTLEMENT OF CADMIUM-PLATED (0.001 IN.) STEEL TUBE (DATA BY MENASCO)⁽¹⁸⁷⁾



20	212 F 302 F Rc 56.5		Befor Platin
0	4 8 12 16 20 24 28 Baking Time, hr	Tensile Strength, psi	234,00
FIGURE 65.	EFFECTS OF TEMPERATURE AND TIME ON RECOVERY FROM HYDRO-	Yield Strength, psi	208,00
	GEN EMBRITTLEMENT OF CADMIUM- PLATED SPRING-STEEL STRIP (0.008 IN. x ¹ / ₄ IN.)(AFTER EAKIN	Elongation, percent in l inch	15.0
	AND LOWNIE, REFERENCE 129) ⁽¹⁸⁷⁾	Reduction in Area,	57 0

hydrogen embrittlement due to electroplating, some tests conducted on SAE 4340 steel indicated that the higher temperatures may be detrimental to the ductility of high-strength cadmium-plated parts. In any event, temperatures over 610 F, the melting point of cadmium, must be avoided, because liquid cadmium can cause cracking of lowhydrogen steel. Also, baking temperatures higher than 400 F produce discoloration of cadmiumplated parts, and thus, are undesirable. It has been suggested that the presence of such an oxide film on the surface of the cadmium plate further interferes with the removal of hydrogen, and this may explain the adverse effects of baking at temperatures above 400 F that sometimes have been observed. Another explanation that might account for such an observed effect is the formation of a brittle layer at the steel-cadmium interface as the result of diffusion of some of the cadmium into the steel.

Before strength levels were boosted to the point where delayed brittle failures were encountered, the most common relief practice for hydrogen embrittlement resulting from cadmium plating consisted of baking at 375 to 400 F for 3 or 4 hours. When hydrogen-induced delayed brittle failures were encountered, baking times often were increased to 23 hours, but this practice was only partially effective in alleviating the susceptibility to hydrogen cracking.

Raring and Rinebolt⁽¹³⁷⁾ studied the effect of baking at 350 F for 11/2 hours on cadmiumplated, vacuum-melted 4340 steel at the 230,000psi strength level. Although the baking treatment restored the breaking strength of the plated specimens in the short-time tensile test nearly to that of the unplated specimens, the susceptibility to delayed brittle failure was not changed appreciably. After plating and baking, the lower critical stress was only about 47 percent of the notched tensile strength of specimens that were not plated. The hydrogen content before baking was 0.7 to 1.1 ppm; after baking it was 0.6 to 0.7 ppm. The effect of the treatment on the short-time tensile properties of unnotched tensile specimens was as

	Before Plating	After Plating and Baking 1½ Hours at 350 F	
Tensile Strength, psi	234,000	230,000	
Yield Strength, psi	208,000	208,000	
Elongation, percent in l inch	15.0	15.0	
Reduction in Area, percent	57.0	52.5	

Various other investigators also showed that, although suitable baking treatments resulted in full recovery of lost ductility, as indicated by a conventional tensile test, frequently such treatments did not eliminate susceptibility to delayed failures, especially for materials of the higher strength levels; for example, see Figure 66. Delayed failures have occurred after baking for times at least as long as 100 hours. Sometimes the static loading tests of notched specimens may indicate essentially no recovery as a result of a baking treatment. Thus, it has been shown conclusively that data obtained in a short-time tensile test are not suitable criteria of the susceptibility to delayed brittle failure. Also, baking does not insure elimination of the susceptibility to such failures.

The recovery of parts cadmium plated on only one surface is considerably faster than that of parts plated on both surfaces. Figure 64 illustrates the difference for hollow parts plated in the one case only on the outer surface and in the other case on both surfaces. Now and then a specific lot

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100

80

60

80

60

40

0

Bend Value, percent vise opening

437F

4

8







of cadmium-plated material is found which is especially resistant to the relief of hydrogen embrittlement. Often this is the result of variations in the structure of the cadmium plate. A somewhat porous plate offers less resistance to the effusion of hydrogen. Methods have been developed to intentionally produce a porous plate which permits hydrogen to diffuse out of the surface in a subsequent bake to remove hydrogen introduced in the plating operation [for example, see Reference (195)]. The factor that influences the type of plate produced is the use of a brightener in the cadmium-plating solution. One investigation showed that very small amounts of such an additive greatly reduced the recovery from hydrogen embrittlement of SAE 4340 steel heat treated to a hardness of Rockwell C 48. Also, some plating baths are more efficient than others and plate out less hydrogen at the metal surface.

Residual, or Nondiffusible, Hydrogen

Studies of the removal of hydrogen from steel led to the concept of "residual", or "nondiffusible", hydrogen in iron and steel. (184, 196, 181, 47, 160)Residual hydrogen has been widely known for many years in the analysis of steels for hydrogen. It is chiefly responsible for the lack of general acceptance in the U. S. of the warm-extraction analytical method for hydrogen determination. The hydrogen is not all removed by this method, which involves measuring the hydrogen evolved by heating in the α -range. Apparently, residual hydrogen can be extracted only at temperatures above about 1470 F (800 C). Different investigators have reported that



FIGURE 67. CURVE OF EXTRACTION OF HYDRO-GEN IN VACUUM AT 1110 F (600 C), SHOWING TWO APPARENTLY DIF-FERENT REGIMES: A, PERIOD OF RAPID EVOLUTION; B, PERIOD OF SLOW EVOLUTION; Q/R. = RATIO OF HYDROGEN EXTRACTED AT ANY TIME TO THE TOTAL HYDROGEN COLLECTED AFTER 40 HOURS [AFTER BLETON, et al⁽¹⁹⁷⁾]⁽¹⁹⁸⁾

(Samples $\frac{1}{2}$ inch diameter and $\frac{1}{2}$ inch long.)

this residual form of hydrogen can be removed from steel only by drastic vacuum treatments at temperatures well up into the austenitic range, or even in the molten state (vacuum-fusion analysis). Thus, the terms "residual" and "nondiffusible" were used to describe a form of hydrogen in steel which is bound far more tenaciously than that described by the usual diffusion coefficient (given in Equation 38, page 40, and Figure 32, page 44, for example).

It was the results of experiments on the rate of removal of hydrogen from steel that led a number of investigators to suggest that this element exists in steel in two forms, each with different effects on the physical properties of steel. This conclusion was based in part on the type of extraction curve shown in Figure 67. However, this hypothesis is not confirmed by vacuumextraction experiments performed at 1200 F (650 C) and a theoretical treatment of diffusion performed by Demarez, Hock, and Meunier. (198) Their experiments purportedly showed that there is reasonable agreement between observed extraction rates and rates predicted from the laws of normal diffusion. They stated that it is thus possible to estimate the course of hydrogen evolution from the usual rolled products if the diffusion coefficient is known.

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Johnson and Hill(10) suggested recently that residual hydrogen is associated chemically with an impurity in iron that results in a compound that decomposes rapidly above 1470 F (800 C). They found large amounts of nondiffusible hydrogen in airmelted iron and only small, but detectable, amounts in several heats of vacuum-melted iron. They first considered that the compound is water resulting from the reduction of FeO by dissolved hydrogen. However, the oxygen contents of all vacuum-melted heats were sufficient to form large quantities of water, but very little residual hydrogen formed, even in a vacuum-melted heat to which an addition of 1200 ppm of oxygen was made. Recent work by Podgurski⁽⁵⁹⁾ suggests that the compound may be methane. The identity of the compound is still uncertain. Hill and Johnson found that the rate of residual-hydrogen formation is very low at temperatures below 930 F (500 C), but increases rapidly at temperatures above 930 F and reaches a maximum at 1110 F (600 C). They suggested that the chemical fixing of hydrogen in steel occurs during cooling between 1470 and 930 F (800 and 500 C). If the steel is quenched through this range, very little should form and nearly all the hydrogen in the material should remain diffusible. If the steel is slowly cooled, extensive conversion of diffusible to residual hydrogen probably occurs. Previously reported contents of residual hydrogen in steel generally did not exceed 1 ppm. However, Hill and Johnson(13) demonstrated that up to 19 ppm residual hydrogen may be formed in air-melted iron by prolonged holding at 1110 F (600 C) under hydrogen-pressure levels corresponding to diffusible hydrogen concentrations of approximately 7 ppm.

Johnson and Hill^(47,13) found that certain grades of iron, when treated for several days at high pressures and at temperatures between 840 and 1200 F (450 and 650 C), will absorb considerably more hydrogen than is indicated by the solubility data shown in Figure 8, page 10. Furthermore, they found that the hydrogen thus introduced could be extracted in two separate fractions, one of which was evolved in accordance with the diffusion coefficient for annealed iron given in Figure 32, page 44. They found that the other fraction was removable only by more extreme extraction conditions, such as heating for several weeks at about 1110 F (600 C) or by heating for about 1 hour at 1560 to 1650 F (850 to 900 C).

In Figure 68, the evolution characteristics of hydrogen from two samples of iron, only one of which contained residual hydrogen, are shown. One sample was from a heat of iron that had been melted in air, and the other was vacuum melted; both were made from electrolytic-iron melting stock. Both samples were soaked in hydrogen gas at 62 atm pressure and 1085 F (585 C) for 5 days. This treatment introduced about 8.2 ppm of hydrogen into the vacuum-melted iron. Evolution of hydrogen from the hydrogen-charged vacuummelted iron was completely normal. All the hydrogen was evolved at 1065 F (575 C) at a rate which

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could be calculated from the known diffusion constant. After 6 hours at 1065 F, the temperature was raised to 1470 F (800 C), but no additional hydrogen was evolved. For the air-melted iron, the rapid initial evolution of "diffusible" hydrogen at 1065 F was followed by a slow evolution which was still in progress at the end of 2 days. It appeared that complete extraction of the hydrogen at this temperature would take several weeks, so the temperature was raised to 1290 F (700 C). This resulted in a marked increase in the rate of evolution. Since it appeared that another day would be required for complete extraction, the temperature was raised again, this time to 1560 F (850 C). Again, the evolution increased, but extraction was complete in 1 hour at this temperature. Subsequent vacuum-fusion analyses of both samples showed that neither retained more than 0.2 ppm of hydrogen after the solid-state extractions just described. Thus, a total of 23.4 ppm of hydrogen was evolved from the air-melted sample, as compared with 8.2 ppm from the vacuum-melted sample, even though identical charging conditions had been used. The value of 8.2 ppm was exactly what was calculated using Sieverts' law and the data in Figure 8, page 10, and, therefore, represents the solubility at 62 atm and 1085 F (585 C). Thus, the hydrogencharged, air-melted sample contained 15.2 ppm more hydrogen than given by the solubility data. Further experimental work showed that this extra hydrogen was evolved with an activation energy 16 times greater than the activation energy for normal diffusion at these extraction temperatures.



FIGURE 68. EVOLUTION OF HYDROGEN FROM CYLINDERS OF IRON 0.75 INCH IN DIAMETER BY 1.25 IN. LONG⁽¹⁶⁷⁾

Three different air-melted heats of iron showed similar tendencies to absorb extra amounts of hydrogen above the values indicated by the solubility curves (Figure 8). The amount of extra hydrogen increased with increasing duration of the charging treatment and with increasing pressure of gaseous hydrogen. No extra hydrogen was found in samples treated at temperatures below 750 F (400 C), and only 2 to 5 ppm of extra hydrogen was found in samples treated at 1200 F (650 C). The largest amount of extra hydrogen found was 18 ppm, and this was in samples treated at 1085 F (585 C) for 150 hours. No detectable amounts of residual hydrogen were formed with 1-hour treatments, only the amount of hydrogen indicated by the solubility curve in Figure 8 being introduced.

Three different heats of vacuum-melted iron all showed a similar ability to resist introduction of extra hydrogen, none being found in any heat after soaking in hydrogen at several temperatures and pressures covering the ranges in which residual hydrogen was introduced into air-melted samples.

By using samples of vacuum-melted iron or by using very short charging periods with air-melted iron, it was possible to prepare samples which contained no residual hydrogen. These precuations were observed in all experiments used by these investigators in determining the solubility and the diffusivity of hydrogen in steel (Figures 32 and 8). Thus, the deviations these investigators found in solubility and diffusivity at low temperatures (as discussed previously in the present report) were not related to residual hydrogen.

It has been shown that hydrogen embrittlement and delayed brittle failure of steel are related to the ability of hydrogen to diffuse, as has been discussed in earlier reports by the present authors (DMIC Memorandum 180 and DMIC Report 196). It is clear from the kinetics of evolution that in whatever form the "residual" hydrogen is contained in iron and steel, it is firmly bound and is essentially immobile and unable to diffuse in the temperature range (near room temperature) where these phenomena are observed. This indicates that residual hydrogen cannot contribute to these effects. Thus, conversion of hydrogen to this form would appear to have possible benefits. It would appear to be possible to convert some of the hydrogen in commercial steels (perhaps excluding vacuum-melted steel) into a chemically bound form during prolonged heat treatments at about 1110 F. However, this is considerably above the tempering temperature used for most high-strength steels, with the exception of those that exhibit secondary hardening. Also, the kinetics of its formation prohibit the formation of this bound hydrogen at all temperatures normally used for electroplating from aqueous solutions and in the postplating baking treatments.

Although it would appear that residual hydrogen cannot contribute to hydrogen embrittlement or hydrogen cracking of steel, Hill and Johnson are of the opinion that residual hydrogen is responsible for serious grain-boundary weakening. Because of these apparently separate effects of the two forms of hydrogen, these investigators suggested that determinations of both diffusible and residual hydrogen should be made in the analysis of steel.

Podgurski⁽⁵⁹⁾ studied the anomalously high solubility reported for hydrogen in cold-worked steel at temperatures above 390 F (200 C). He attributed this to methane formed by a reaction between hydrogen and cementite; he quantitatively isolated this methane from steel.

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APPENDIX E

SAMPLE EXAMINATION Hillman-Kelley Global Marine, Inc. 5" O.D. x 19.50[#] Special BR-135 Drill Pipe

all

10: r. D. Derve Works Chief Metallurgist

FROM: Tube Investigation Division

SUBJECT: <u>SAMPLE EXAMINATION</u> Hillman-Kelley Gloval Marine, Inc. 5" O.D. x 19.50# Special BR-135 Drill Pipe

Hillman-Kelley forwarded a 8-1/2" long pipe sample split in half of J&L 5" O.D. x 19.50# Special BR-135 drill pipe submitted to them by Global Marine, Inc. Global Marine, Inc. requested the following opinions on the drill pipe that they are using aboard the Glomar Challenger:

- (1) What is the scale?
- (2) What is its cause?
- (3) What is the best methods of removal both on shipboard at a remote location and at Houston?
- (4) What is the best corrective action?
- (5) What is the best method to prevent scale formation and pitting?

Sample Examination: (Received 9/25/70):

Visual

Macroscopic examination revealed severe general corrosion pitting on the inside pipe surface. There was no transverse fatigue cracking noted. Refer to attached photograph. The outside surface did not exhibit corrosive attack.

Longitudinal Mechanical Properties:

Specimen length and inside surface condition precluded determination of mechanical properties.

Hardness Determination:

33.3 Rockwell "C" Scale (average of three readings)

Chemical Check Analysis (Base Metal):

С	Min	P	S	Si	Mo
.26	1.67	.018	.020	.17	.44

Chemical Analysis - Corrosion Products:

<u>C1 S03 C03</u> 0.21% 0.55% 1.39%

Wall Thickness:	Sample *	Nominal	Minimum
	.358/.375"	.362"	.317"

* Apparent original wall thickness

Microexamination: Structure was normal for Special BR-135 drill pipe, tempered /249

Page 2 November 5, 1970 SAMPLE EXAMINATION Hillman-Kelley

Discussion:

Chemical check analysis of the base metal, wall thickness, hardness, and microstructure were normal for 5" O.D. x 19.50# Special BR-135 drill pipe.

The inside surface of the pipe sample exhibited severe general pitting typical of environmental corrosive attack. Chemical analysis of the scale (corrosion products) indicated .21% chlorides, .55% sulfites, and 1.39% carbonates. While percentage figures for chlorides, sulfites, and carbonates appear low, they are indicative of a corrosive media in the drilling string.

The small concentration of chloride indicates that salt water was not a major contributor but enhanced oxidation corrosion by serving as an electrolyte. The scale was magnetic which indicates the presence of a substantial amount of Fe_3O_4 . This is typical for a string filled with drilling mud which limits the amount of available oxygen.

The presence of sulfites indicates the presence of hydrogen sulfide in the system which reacts with steel in the presence of water to form iron sulfide. This sets up a galvanic cell with the steel serving as a sacrificial anode resulting in pitting.

The concentration of carbonate in the scale indicates that excessive CO_2 was in the system and this was the most severe causative mechanism of corrosive attack. The carbon dioxide converting to free carbonic acid caused severe pitting.

Removal of the scale can be accomplished by abrasive blasting (shot or sand). Acid pickling after abrasive blasting of high strength-heat treated material is not recommended because of the potential of hydrogen embrittlement. Page 3 November 6, 1970 SAMPLE EXAMINATION Hillman-Kelley

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Cleaning the scale off the pipe will not significantly increase the life of the drill pipe because of the advanced stage of corrosion pitting. Continued use of drill pipe with the inside surface condition exhibited by the sample will ultimately result in failure by initiation and propagation of fatigue cracks at the base of corrosion pits.

The use of appropriate inhibitors is recommended. The services of a corrosion engineer should be obtained to determine the proper inhibitors and concentrations required to prevent further deterioration of the drill pipe.

C. a. Juess C. A. Guess

C. A. Guess Aliquippa Metallurgical Department

2/16/10

CC: Tech.Ser.Div.-9 A.R.Aynardi R.L.Guernsey J.J.Lenker.-2 C.J.Millner E.J.Newton

CAG:1m

HILLMAN - KELLEY Global Marins, Inc.

PHOTOGRAPH



General corrosion pitting observed on the inside surface. No transverse fatigue cracking was noted.

APPENDIX F

D

Study of Hydrogen–Stress Cracking of Deep Sea Drilling Project S–135 Drill Pipe if Exposed to Black Sea Water

STUDY OF HYDROGEN-STRESS CRACKING OF DSDP S-135 DRILL PIPE IF EXPOSED TO BLACK SEA WATER

ABSTRACT

This study reviews the limitations of operating the Deep Sea Drilling Project's S-135 drill pipe in sea water containing hydrogen sulphide.

INTRODUCTION

The design and selection of the DSDP drill pipe did not preclude its use in sea water other than that in deep ocean basins. The extension of the DSDP program is producing requests to drill in waters that contain varying amounts of H₂S. As many steels are susceptible to hydrogen stress cracking if exposed to H₂S, DSDP has reviewed available literature and discussed the problem with experts in this field.

CONCLUSIONS

- A review of papers on hydrogen-stress cracking of steels and discussions with people knowledgeable in this field indicate the S-135 drill pipe is susceptible to hydrogenstress cracking.
- The DSDP S-135 drill pipe apparently is in the transition zone between high strength steels very susceptible to hydrogen stress cracking and low strength steels which are not affected by hydrogen entering the metal lattices.
- 3. The field studies on the effect of hydrogen sulphide on oil field tubular goods has been directed to casing and tubular goods, so information on its effect on drill pipe of grades above "E" is meager.

- 4. So many factors influence hydrogen-stress cracking, e.g., steel hardness, chemical content, temperature, length of exposure to contamination, stress level, that physical tests of a given steel in various environment are necessary to determine its susceptibility to hydrogen stress cracking.
- 5. Preventing the entrance of hydrogen into the steel reduces hydrogen-stress cracking. The effect of the zinc rich coating on the DSDP drill pipe in preventing hydrogen entrance into the S-135 pipe is unknown.
- 6. Sonoscope inspections of drill pipe (tuboscope) have indicated there have been no drill string failures introduced by excessive stress levels in the drill pipe operating in a corrosive environment (sea water).

RECOMMENDATIONS

- Contact testing laboratories such as Battelle Memorial Institute and Southwest Research for a recommended program to determine if DSDP's S-135 drill pipe is susceptible to hydrogen stress cracking and if it is, what is the maximum safe stress level at which we can operate the pipe in sea water with H₂S concentrations such as in the Black Sea.
- 2. From the above report, establish operating limits for the S-135 drill pipe.

DISCUSSION

The failures in the oil industry of steel tubular goods attributed to hydrogen stress cracking from H₂S environment has established two rules of thumb to select steels:

 An elongation of 23% to 25%. This should be after a quench and temper, as as-rolled goods usually have local hard spots. The above valve has been established by the Mannersman Co.⁽¹⁾ From experience gained in Canadian Oil Fields with high H₂S concentrations.

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Keep yield point of steel below 80,000 psi and hardness below Rc 22. These values (2,3,4,) are generally accepted by the oil industry and have been established by experience of Pan American (and others) in West Texas gas fields with high H₂S concentrations.
 In addition, drilling or production fluids can be treated to reduce the effect of H₂S on steel.

Normally, the drilling industry can minimize failures of hydrogen stress corrosion by using drill pipe that meets with one and two above. This sometimes requires drill strings tapered in weight and/or grade.

The ability of the Challenger to cut and recover a 2 1/2" dia core through 22,500' of drill string requires a drill string of high tensile strength. The pipe racker configuration requires a string of uniform weight and grade. Physical limitations of the ship make it difficult to carry two different strings, or would require off-loading the standby string. As drilling and coring is done without drilling fluid to the surface, treating the drilling fluid is very costly. Thus, if the DSDP drill pipe is to be used to drill in the Black Sea, it should be assumed it will be exposed to the H₂S at all times.

As the Glomar Challenger operates without a supply ship, logistically it would be easier to drill the holes in the Black Sea with the S-135 drill pipe. It should be noted the hydrogen in the steel is released if the pipe is exposed to the atmosphere, so is not permanently changed.

As the drill string length to drill in the Black Sea would be slightly over half the length of the longest drill string used, the pipe could loose an appreciable amount of its tensile strength and still operate satisfactorily. Thus, pending actual physical testing, use of the S-135 pipe in the Black Sea seems possible.

Below are excerpts from Battels Defense Metals Report on "Stress-Corrosion Cracking and

Hydrogen-Stress Cracking of High-Strength Steel."

Phosphorus, arsenic, antimony, sulfur, selenium, tellurium, and apparently the cyanide (CN) ion are among the cathodic poisons which promote the entry of hydrogen into iron or steel during pickling or cathodic charging. (Hydrogen sulfide is a potent cathodic poison.

All types of steels with hardnesses above about Rockwell C 23 will crack in aqueous solutions that contain hydrogen sulfide. The results McGlasson and Greathouse obtained for a number of steels stressed below the yield point are listed in Table 1. These investigators also have summarized the work of other investigators on steels stressed above the yield stress (that is, cold worked) either before or during the tests (Table 2). They concluded that for a given strength level, low-chemistry quenched-and-tempered steels are less susceptible to sulfide cracking than are more highly alloyed steels. Additions of manganese, molybdenum, or nickel reportedly increase the susceptibility to sulfide cracking, while chromium tends to reduce susceptibility. Also, it is claimed that increasing the carbon content for the same strength level improves resistance to sulfide cracking. The French have developed a steel with a yield strength in the 80,000 to 90,000-psi range that is superior to N-80 steel in resistance to sulfide cracking. It has the following composition:

С	0.12 percent	
Cr	2.0 percent	
Mo	0.3 percent	
AI	0.3 percent	

For resistance to hydrogen sulfide cracking in oil-field applications such as casings and drill pipe, it is customary to specify that the steel be no harder than rockwell C 22 (ultimate tensile strength of about 110,000 to 115,000 psi and yield strength on the order of 95,000 psi, depending upon the structure). However, failures by sulfide cracking can occur even at that specified maximum strength level if the service stress is unusually high, or at even lower levels if the steel is plastically deformed. In the laboratory, McGlasson and Greathouse found no cracking on exposure to hydrogen sulfide solutions for steels with a hardness of Rockwell C 23 or less. However, in other work, sulfide cracking reportedly was obtained in material with yield strength of 67,000 psi at an applied stress of 55,000 psi, but steels with yield strengths in the range of 64,000 to 67,000 psi survived 300 days without cracking. Thus, there are hardness and applied-stress minimums below which a given steel will not crack when exposed to an aqueous system containing hydrogen sulfide; these are inversely related. The effect of hardness on the susceptibility to sulfide cracking of heat-treated AISI 4140 bolts in an investigation by Warren and Beckman

D. L. Sims 5/25/70

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cc K. E. Brunot W. R. Jack V. F. Larson M.N.A. Peterson

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BLACK SEA H2S (MILLI MOLES PER LITER)



Chemical Composition and Mchanical Properties

GLOBAL MARINE - 5" x 19.50%, S-135 53"x 24.70%, S-135

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	· · · ·									
+	Heat		() •)				La	die Ana	lysis	8
L	Nutaber	Yield	Tensile	% EL-	<u>c</u>	Mm	<u>P</u> .	S	Si	Mo
	619970	144,630	161,270	18.0	.26	1.66	.021	.020	.18	.45
	S.R.	149,080	162,600	20.0						
	610017	149,830	162,080	20.0	.23	1.82	.022	.019	.22	.48
	S.R.	144,060	157,360	19.0	•	. s S.				
	610111	142,370	156,560	21.0	.26	1.71	.022	.020	.19	.45
	S.R.	142,970	156,150 .	18.0			·			
	628516	144,740	162,520	19.0	.28	1.73	.024	.020	.18	.55
	S.R.	153,550	166,400	24.0						• •
	628553	136,640	153,720	22.0	.26	1.66	.017	.019	.18	.48
	S.R.	143,570	155,320	18.0	5			1. N. A	· · · · · · · · · · · · · · · · · · ·	
	637859	144,740	159,110	18.0	.27	1.82	.025	.021	.16	.48
	S.R.	151,570	164,540	20.0	eri Qiri Al					
	637992	141,900	158,850	16.0	.27	1.74	.020	.020	.23	.43
	S.R.	143,330	155,500	22.0						•
	554702	148,470	160,280	18.0	.26	1.66	.021	.020	.18	.45
	S.R.	147,730	160,430	21.0			e .			
	554703	145,300	160,950	20.0	.28	1.82	022	.019	.22	:48
	S.R.	143,960	158,950	20.0			4			
	610065	136,620	153,670	21.0	.28	1.70	.022	.019	.18	.48
										S

APPENDIX G

Stress-Corrosion Cracking and Hydrogen-Stress Cracking of High-Strength Steel

DMIC Report 232 July 29, 1966

STRESS-CORROSION CRACKING AND HYDROGEN-STRESS CRACKING OF HIGH-STRENGTH STEEL

DEFENSE METALS INFORMATION CENTER

Battelle Memorial Institute Columbus, Ohio 43201

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- 1. To collect, store, and disseminate technical information on the current status of research and development of the above materials.
- 2. To supplement established Service activities in providing technical advisory services to producers, melters, and fabricators of the above materials, and to designers and fabricators of military equipment containing these materials.
- 3. To assist the Government agencies and their contractors in developing technical data required for preparation of specifications for the above materials.
- 4. On assignment, to conduct surveys, or laboratory research investigations, mainly of a short-range nature, as required, to ascertain causes of troubles encountered by fabricators, or to fill minor gaps in established research programs.

Contract No. AF 33(615)-3408

Roger J. Runck Roger J. Runck

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STRESS-CORROSION CRACKING AND HYDROGEN-STRESS CRACKING OF HIGH-STRENGTH STEEL

E. E. Fletcher, W. E. Berry, and A. R. Elsea*

SUMMARY

Any of the high-strength steels is susceptible to delayed cracking under suitable conditions of applied or residual stress and environment. Frequently, such a brittle failure occurs at a stress that is only a fraction of the nominal yield strength of the material. Considerable controversy exists over whether such failures result from two separate and distinct phenomena or whether there is but one mechanism called by two different names (hydrogenstress cracking and stress-corrosion cracking). The appearance of the fracture surface is similar for both modes of failure, and this is one reason for difficulty in distinguishing between the two mechanisms and thereby contributes to the controversy.

This report has been prepared with the purpose of pointing out wherein the two fracture mechanisms under consideration are similar and wherein they differ. From the evidence available today, the present authors have concluded that there are two distinct mechanisms of delayed failure.

Stress-corrosion cracking is the process in which a crack propagates at least partially by the stress-induced corrosion of a susceptible metal at the advancing tip of the stress-corrosion crack. There is considerable evidence that this cracking results from the electrochemical corrosion of a metal subjected to tensile stresses, either residual (internal, such as may result from welding or heat treatment) or externally applied.

The phenomenon of hydrogen-stress cracking is cracking which occurs as the result of hydrogen in the metal lattice in combination with tensile stresses (or stresses that have tensile components). Hydrogen-stress cracking cannot occur if hydrogen is prevented from entering the steel, or if hydrogen that has entered during processing or service is removed before permanent damage has occurred. It generally is agreed that corrosion (the dissolution of metal with the formation of a corrosion product or products) plays no part in the actual fracture mechanism. It is possible that under some circumstances both types of cracking occur simultaneously; this situation would be very confusing to those attempting to study the mechanism of one or the other of these types of cracking.

It is noteworthy that both fracture mechanisms seem to be promoted by higher strength levels, both require tensile stresses, both seem to involve an interval of time for the nucleation of a crack, and the fracture mode of both is generally intergranular, although some transgranular cracking is encountered. Another similarity is that, during or after exposure to certain environments, alloys fail under stresses it was assumed they could bear. Furthermore, both involve events occurring on the atomic scale.

However, certain differences also have been reported. For example, hydrogen-stress cracking is promoted by cathodic conditions, while stresscorrosion cracking is promoted by anodic conditions. Generally, increasing the temperature increases the likelihood of stress-corrosion cracking, but hydrogen-stress cracking is more prone to occur at temperatures not far from room temperature.

Insofar as is now known, the important factors in these two types of delayed cracking are strength level, steel composition, steel structure, applied and residual stresses, environment (whereby there is a tendency toward corrosion or the introduction of hydrogen into the steel), and time. These factors have been considered individually.

Strength level seems to play a different role in the two mechanisms. For example, nitrate and caustic cracking (stress-corrosion cracking) commonly occur in steels with yield strengths considerably below the lowest level in which hydrogenstress cracking can be readily achieved. However, a material is more prone to fail by either mechanism as the strength level is increased. Not enough data are available regarding cracking behavior for different steel compositions to shed much light on the issue. However, the cracking behavior of different steel structures provides limited evidence that these are separate phenomena. Applied stress apparently has the same role in both mechanisms.

^{*} Senior Metallurgical Engineer, Associate Chief, and Associate Chief, respectively, Ferrous and High Alloy Metallurgy and Corrosion Research Divisions, Battelle Memorial Institute, Columbus, Ohio

Some aspects about the environment provide perhaps the most convincing evidence for two mechanisms. Delayed fractures occurring under conditions of complete cathodic protection can hardly be attributed to stress corrosion. Conversely, fractures occurring under conditions where the cathode reaction does not involve reduction of hydrogen ions (as for example in nitrate solutions) can hardly be hydrogen-stress cracking. Under some circumstances, the mode of failure is not clearcut, as in hydrogen sulfide solutions or water. It appears likely that both types of cracking can be associated with some of these environments and, furthermore, that both may be acting simultaneously under some circumstances.

If impressing a low level of cathodic current (which tends to polarize the local cathode areas in the direction of the potential of the anodes) very greatly extends the breaking time, one concludes that the cracking process in the absence of impressed current is properly designated stresscorrosion cracking. However, if, for a given medium, imposing a small cathodic current decreases the breaking time whereas imposing anodic currents greatly extends life, the behavior indicates that in the case of no impressed current (the "natural" conditions), the failure occurs by hydrogen-stress cracking.

These responses to impressed currents appear to offer a promising means of distinguishing between the two mechanisms, at least in the laboratory. However, they probably do not afford a practicable means of mitigating against cracking in highstrength steels. For example, if a system is made a little too cathodic in order to combat stresscorrosion cracking, hydrogen cathodically charged into the steel may well cause delayed failures. Likewise, anodic currents impressed on a system to avoid hydrogen-stress cracking may accelerate ordinary corrosion beyond acceptable limits.

The nature of the potential-time curves obtained during stress-corrosion cracking and hydrogen-stress cracking, respectively, indicates a difference between the two fracture mechanisms.

A different response to variations in the temperature of the environment also suggests two modes of delayed brittle fracture.

It seems to be rather generally regarded that dissolution of metal (corrosion) at the advancing tip of the crack is essential for stresscorrosion cracking, whereas corrosion plays no part in the hydrogen-stress-cracking mechanism. However, corrosion can and frequently does involve liberation of atomic hydrogen as the cathode reaction, and a portion of this hydrogen may enter the steel and cause hydrogen-stress cracking.

INTRODUCTION

This report was prepared in response to a request from the Working Panel on Metals, Sub-Group P on Materials, of The Technical Cooperation Program. The request was authorized by a letter from N. E. Promisel to the U. S. Department of Defense on June 17, 1965.

It is the purpose of this review to point out wherein the two fracture mechanisms under consideration (stress-corrosion cracking and hydrogenstress cracking) are similar and wherein they differ. To accomplish this end, the important factors that influence the tendency to promote cracking by the two mechanisms will be reviewed insofar as information is generally available in the current technical literature. The important factors as now known are strength level, steel composition, steel structure, applied and residual stresses, environment (whereby there is a tendency toward corrosion or the introduction of hydrogen into the steel), and time. These factors will be considered one at a time.

The report describes the current thinking and the state of the art with regard to these two modes of fracture. The present discussion will be limited to high-strength steels except where it is necessary to bring in other materials to point out differences between the two phenomena.

FACTORS AFFECTING STRESS-CORROSION AND HYDROGEN-STRESS CRACKING

Introduction

Hydrogen-stress cracking and stresscorrosion cracking are two fracture mechanisms that have been responsible for many serious and unexpected failures in high-strength steel parts (and in a number of other structural metals as well). During or after exposure to certain environments, alloys have failed under applied stresses which it was assumed they could safely withstand. Much effort has been expended in determining the relative sensitivity of various structural metals to failure by these mechanisms and in evaluating the effects of various environments on the tendency toward failure. In addition, many research investigations have been directed toward determining the mechanisms by which cracks are initiated and propagated in hydrogen-stress cracking and stress-corrosion cracking. In spite of this effort, neither mechanism is completely understood at present. The different investigators seem to have had varying ideas about each mechanism, and a number of theories have been proposed for each. For heither mechanism has any one theory been universally accepted as providing the explanation of how the environment to which the alloy is exposed can initiate a fracture and enable it to propagate, frequently under a very small stress. It has been reported that both stress-corrosion-cracking and hydrogen-stress-cracking failures have occurred in the laboratory at stresses as low as one-twentieth of the yield strength.

Because the two fracture mechanisms are so much alike, some investigators believe that they are one and the same, the only difference being in the names used to describe the phenomenon. However, other investigators claim that, although there are some apparent similarities, the basic mechanisms are different, and many of the limiting conditions for the occurrence of the two types of cracking are different. A failure will sometimes occur under conditions such that it is called stress-corrosion cracking; at other times the conditions are such that the failure is called hydrogen-stress cracking. Thus, there may be two distinct fracture mechanisms. However, there is incomplete agreement as to which cracking phenomena should be termed stress-corrosion cracking and which are actually hydrogen-stress cracking.

In general, the two mechanisms may be characterized as follows:

Stress-corrosion cracking is the process in which a crack propagates at least partially by the stress-induced corrosion of a susceptible metal at the advancing tip of the stress-corrosion crack. There is considerable evidence that this cracking results from the electrochemical corrosion of a metal subjected to tensile stresses, either residual (internal, such as may result from welding or heat treatment) or externally applied. That is, the propagation of a stress-corrosion crack depends on the flow of current between localized anodic and cathodic areas. Frequently, the corrosive environment alone will not cause corrosion, but only does so in the presence of stress. Very little evidence of corrosion may be apparent, perhaps none at all to the unaided eye. Indeed, if corrosion occurs uniformly at a high rate, stress-corrosion cracking is not encountered.

Cracking involves the creation of new surfaces, which requires energy; but of course, cracking is not the only way in which new surfaces are generated. Failure by dissolution of grain boundaries through selective electrochemical attack is not true fracture or cracking, for fracture involves the breaking of metal bonds. That is, fracture is a cohesive separation which occurs at a high speed (a maximum of about half the elasticshear-wave velocity and frequently about one-third the speed of sound), although the fracture may grow in small discrete increments, each propagating at high speed. Cracking occurs with little evidence of ductility, that is, with little plastic deformation and, hence, failures by cracking are termed brittle failures. Shear failures, on the other hand, occur by relatively slow propagation involving crystallographic slip and considerable plastic deformation; these are ductile failures.

A number of different phenomena have been called hydrogen embrittlement. The present report is concerned only with the one known as hydrogeninduced delayed brittle failure, hydrogenembrittlement cracking, or hydrogen-stress cracking. The phenomenon of <u>hydrogen-stress cracking</u> is cracking which occurs as the result of hydrogen

in the metal lattice in combination with tensile stresses (or stresses that have tensile components). Hydrogen-stress cracking cannot occur if hydrogen is prevented from entering the steel, or if hydrogen that has entered during processing or service is removed before permanent damage has occurred. A critical combination of hydrogen and stress must be present in a susceptible material at a location suitable for crack nucleation, and under conditions that permit the hydrogen to diffuse through the lattice, before hydrogen-stress cracking will occur. It generally is agreed that corrosion (the dissolution of metal with the formation of a corrosion product or products) plays no part in the actual fracture mechanism. However, corrosion frequently is accompanied by the liberation of atomic hydrogen as the cathode reaction. If a portion of the atomic hydrogen thus liberated enters the steel, hydrogen-stress cracking then may ensue.

It appears likely that, under some circumstances, both types of cracking occur simultaneously; such occurrences probably are responsible for much of the confusion concerning these two types of cracking.

Insofar as is now known, the important factors that tend to promote cracking by the two mechanisms are strength level, steel composition, steel structure, applied and residual stresses, environment (whereby there is a tendency toward corrosion or the introduction of hydrogen into the steel), and time. These factors will now be considered individually.

Strength Level

Many experiments have shown conclusively that both the minimum stress and the time required to produce hydrogen-stress cracking decrease as the nominal tensile strength of the steel is increased. Thus, these delayed-type brittle failures become an increasingly severe problem as the strength level of steel is increased.

In the laboratory, Slaughter et al^{(1)*} studied the effect of strength level with a group of SAE 4340 steel specimens in which the ultimate tensile strength was varied from approximately 300,000 to 142,000 psi. All of these specimens were fully quenched to produce martensitic structures, and then tempered at temperatures in the range from 300 to 1200 F to produce the desired variations in strength. Smooth (unnotched) specimens were continuously charged with hydrogen cathodically while under sustained load. The results are shown in Figures 1 and 2. In addition, both figures show the effect of differences in structure, but this subject will be discussed later. As the strength level was decreased from 300,000 to 142,000 psi, the time to rupture in the higher range of stress increased by a factor of approximately 100; the stress required to cause rupture in 10,000 minutes increased from 15,000 to 45,000 psi as the strength level was decreased in that range (see Figure 1).

*References are given on pages 19 and 20.

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The results obtained show that hydrogen entering the steel while it is under stress can cause it to lose up to 95 percent of its ability to withstand a sustained load, in the case of steel heat treated to a high strength level. Even at the lowest strength level tested for the tempered martensite (142,000 psi), the steel lost approximately two-thirds of its load-carrying ability.

The same steel was isothermally transformed at 1200 F to produce a pearlitic structure which had a tensile strength of approximately 100,000 psi. Under the same charging and test conditions, delayed failures were obtained in unnotched specimens at this strength level, as indicated by the appropriate curve in Figure 1 and the data point for pearlite in Figure 2.

Other experiments in the authors' laboratory have resulted in hydrogen-stress cracking of two high-strength low-alloy steels with yield strengths in the range of 100,000 to 110,000 psi (achieved by quenching and tempering). Delayed failures of another type of steel with a yield strength of 72,000 psi and tensile strength of 94,000 psi were obtained at three levels of applied stress between 60,000 and 68,000 psi.

Apparently, no well-authenticated instances of hydrogen-stress cracking have been reported for tensile strengths under about 100,000 psi or yield strengths under about 80,000 psi.

In general, field failures by hydrogenstress cracking have occurred only in steels with ultimate tensile strengths above about 160,000 psi. When tensile strengths were increased above about 180,000 psi, considerable trouble was experienced with hydrogen-stress cracking of cadmium-plated aircraft parts, hydrogen having been introduced into the steel during processing or service. There is some evidence that field failures have occurred by hydrogen-stress cracking in steels with tensile strengths as low as 130,000 psi. This conclusion is based on the examination of failures that initiated in hard spots in cathodically protected linepipe.

Unless specifically stated otherwise, when the concept of relative sensitivity to hydrogenstress cracking is considered in this report, reference is made to the effect of the variable under consideration on the minimum stress for failure or on the time to failure for a given material. Any one of the curves in Figure 1 represents the hydrogen-stress-cracking behavior of one specific material under one set of charging conditions. The curve tends to become horizontal for the lower applied stresses (longer failure times). Any combination of applied stress and time under stress that falls below this lower critical stress apparently does not result in failure, at least not in times near the arbitrarily selected runout time. Thus, under the conditions represented by the area below the horizontal portion of the curve (lower critical stress), the steel is considered not to be susceptible to failure. The portion of the figure above









* Battelle Charging Condition A: Electrolyte: 4 per

4 percent by weight of H2SO4 in water

5 drops per liter of cathodic poison composed of 2 grams phosphorus in 40 milliliters of CS₂ 8 ma/in.²

Current density:

Poison:

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the curve indicates the combinations of applied stress and time that will result in failure. Under these applied stresses, the steel is considered to be susceptible to failure. The sensitivity to hydrogen depends on the value of the lower critical stress and on the times to failure at somewhat higher stresses. Thus, any variable which raises the lower critical stress or which moves the curve appreciably to the right (toward longer failure times) may be said to reduce the sensitivity to hydrogen-stress cracking. Conversely, any variable which moves the curve downward or to the left increases the sensitivity.

Stress-corrosion cracking has been encountered in steels of all strength levels. For example, caustic cracking has occurred at elevated temperatures in structural steels with minimum yield strengths of 30,000 to 33,000 psi. A welded storage tank of structural steel in the hot-rolled and annealed condition failed by stress-corrosion cracking after 3 years in service with 50 percent caustic at 120 F. Another tank in service with 50 percent caustic at 100 F failed by caustic cracking in 10 years.

In the laboratory, steels with yield strengths as low as 43,000 psi have failed during exposure to nitrate solutions.

For resistance to hydrogen sulfide cracking in oil-field applications such as casings and drill pipe, it is customary to specify that the steel be no harder than Rockwell C 22 (ultimate tensile strength of about 110,000 to 115,000 psi and yield strength on the order of 95,000 psi, depending upon the structure). However, failures by sulfide cracking can occur even at that specified maximum strength level if the service stress is unusually high, or at even lower levels if the steel is plastically deformed. In the laboratory, McGlasson and Greathouse⁽²⁾ found no cracking on exposure to hydrogen sulfide solutions for steels with a hardness of Rockwell C 23 or less (see Figure 3). However, in other work, ⁽³⁾ sulfide cracking reportedly was obtained in material with a yield strength of 67,000 psi at an applied stress of 55,000 psi, but steels with yield strengths in the range of 64,000 to 67,000 psi survived 300 days without cracking. Thus, there are hardness and applied-stress minimums below which a given steel will not crack when exposed to an aqueous system containing hydrogen sulfide; these are inversely related. The effect of hardness on the susceptibility to sulfide cracking of heat-treated AISI 4140 bolts in an investigation by Warren and Beckman(4) is shown in Figure 4.

Suss⁽⁵⁾ studied stress-corrosion cracking in 300 F water of Type 410 stainless steel heat treated to various strength levels. Failures were obtained for hardnesses above Rockwell C 25 to 30.

As will be discussed more fully in the section on environment, considerable controversy exists as to whether sulfide cracking is a stress-corrosioncracking phenomenon or whether it is a manifestation of hydrogen-stress cracking. In any event, higher strength levels are required for sulfide cracking than for stress-corrosion cracking in other environments. Also, in sulfide cracking, increasing the strength level of high-strength materials causes them to be more susceptible to cracking, as for example with 17-4PH and Type 410 stainless steels. This is the same behavior observed for authenticated hydrogen-stress cracking.

In summarizing the effects of strength level, the controversial area of sulfide cracking will be ignored. On this basis, there is a difference in behavior between hydrogen-stress cracking and accepted types of stress-corrosion cracking, such as nitrate or caustic cracking. This observation suggests that the two phenomena occur by different mechanisms.

Composition

Apparently, all the high-strength steels, including maraging steels and the precipitationhardening stainless steels as well as the martensitic steels, are susceptible to stress-corrosion cracking. (6) Austenitic stainless steels also are susceptible.

All types of steels with hardnesses above about Rockwell C 23 will crack in aqueous solutions that contain hydrogen sulfide. The results McGlasson and Greathouse⁽²⁾ obtained for a number of steels stressed below the yield point are listed in Table 1. These investigators also have summarized the work of other investigators on steels stressed above the yield stress (that is, cold worked) either before or during the tests (Table 2). They concluded that for a given strength level,

We chemistry quenched-and-tempered steels are less susceptible to sulfide cracking than are more highly alloyed steels. Additions of manganese, molybdenum, or nickel reportedly increase the susceptibility to sulfide cracking, while chromium tends to reduce susceptibility. (2,7) Also, it is claimed that increasing the carbon content for the same strength level improves resistance to sulfide cracking. (7) The French have developed a steel with a yield strength in the 80,000 to 90,000-psi range that is superior to N-80 steel in resistance to sulfide cracking. ⁽⁸⁾ It has the following composition:

C	0.12	percent
Cr	2.0	percent
Mo	0.3	percent
A1	0.3	percent.

If the strength levels are the same, a steel of higher carbon content is more resistant to stress-corrosion cracking in nitrate solutions than is one of lower carbon content. In one investigation, nitrate cracking was obtained in a steel with 0. 17 percent carbon, while no cracking



FIGURE 3. HARDNESS VERSUS TIME-TO-FAILURE DATA OBTAINED IN TESTS OF HEAT-TREATED J-55 AND N-80 STEEL IN 5 PERCENT NaCl SATURATED (3000 ppm) WITH H₂S⁽²⁾

Insufficient data to cover entire hardness range. Data illustrate that limiting hardness level depends on applied stress.



FIGURE 4. EFFECT OF HARDNESS ON TIME TO FAILURE OF AISI 4140 STEEL BOLTS EXPOSED IN AN H₂S-H₂O SYSTEM AT 40 C (104 F) AND 250 PSI⁽⁴⁾

	Hardness,	
Material	Rockwell C	Times to Failure ^(a)
9 percent chrome	33	Within 6 months
AISI 4145	30	Within 6 months
N-80(b)	41	2 to 41 minutes
N-80	29	14 to 40 hours
N-80	26	Less than 18 hours
N-80	35	2 to 20 hours
Martensitic stainless	18	Less than 24 hours
9 percent chrome	37	Less than 12 hours

TABLE 1. SULFIDE CORROSION CRACKING PRODUCED IN NOTCHED SPECIMENS LOADED BELOW THE YIELD POINT⁽²⁾

(a) Times to failure vary, depending on applied stress. The nature of the test was such as to not permit definition of exact time to failure.

(b) The series of N-80 tests listed here was made on N-80 material heat treated in ways to provide different hardness levels. This is a Mn-Mo steel containing about 0.46 percent carbon.

TABLE 2. STEEL ALLOYS, HEAT TREATMENTS, HARDNESS, AND STRESSES RESULTING IN SULFIDE CRACKING^{(a)(2)}

Steel	Heat Treatment	Minimum Hardness, Rockwell C	Minimum Applied Stress, psi
API J-55(b)	Quenched and Tempered	30(e)	80,000 ^(c)
API N-80	Normalized Normalized Quenched and Tempered	24 21 28(e)	64,000(c) 120,000(d) 120,000(d)
5 percent nickel 9 percent nickel 5 percent chromium 0 percent chromium	Normalized and Tempered Normalized and Tempered Normalized and Tempered Normalized and Tempered	24 24 30(e) 25(e)	80,000(c) 60,000(c) 120,000(d) 120,000(c)
Type 322 stainless steel (Stainless W)	(f)	44(e)	54,000 ^(c)
Martensitic stainless		₂₅ (e)	100,000(c)

(a) Steels were stressed above the yield stress (cold worked) either before or during test.

(b) Not normal J-55 which is normalized to much lower hardnesses and does not normally exhibit cracking phenomenon.

(c) Specimens were prestressed above yield stress, then subsequently stressed to levels indicated.

(d) Specimens were stressed to a calculated stress which was in all cases above the yield stress of the material.

(e) Insufficient data available to determine these as minimum hardnesses, but they are the lowest hardnesses for which test data are reported.

(f) Yield-strength (129,000 psi) and hardness values suggest that this was in the annealed condition (not age hardened). 7

occurred in steels with 0.24 percent carbon or more. (9) For cracking in 500-hour tests in nitrate, Parkins⁽¹⁰⁾ listed the following values for threshold stress as a function of carbon content (no values of mechanical properties were given):

Carbon Content, percent	Threshold Stress, psi
0.07	5,000
0.08	14,000
0.09	12,000, 18,000
0.11	23,000
0.16	24,000

Uhlig and Sava⁽¹¹⁾ referred to work by Nehl and Werner⁽¹²⁾ which showed that an increase in carbon and aluminum improves resistance to stresscorrosion cracking in boiling 60 percent Ca(NO3) 2-3 percent NH4NO3 solution. Preferably, the carbon and aluminum contents should be not less than 0.15 and 0.04 percent, respectively. They also mentioned work by Athavale and Eilender⁽¹³⁾ which indicated that nitriding a steel produces increased resistance to stress-corrosion cracking. However, other investigators(14) ascribed this improved resistance to compressive stresses introduced in the nitrided surface and not to the nitrogen itself. On the other hand, Uhlig and Sava stated that Parkins(10,15) disagreed with the latter explanation, reporting instead that a nitrided steel that contained 0.095 percent nitrogen and 0.015 percent carbon still did not crack after heat treatment at 900 C (1650 F) followed by furnace cooling to effectively relieve internal stresses. Other investigators(16, 17) have reported that aluminumkilled steels are more resistant to cracking in caustic and nitrate solutions than are similar steels made by other melting practice. Weir⁽¹⁸⁾ reported that a fine-grain steel is more resistant to caustic cracking than is a coarse-grain steel of the same carbon content, and that a fine-grain steel slightly enriched in carbon is quite resistant.

In experiments with 1:1 HCl-water + SeO₂, Fontana⁽¹⁹⁾ observed that cracks started at pits associated with manganese sulfide inclusions.

Although austenitic stainless steels are susceptible to stress-corrosion cracking, it is reported that such a steel with 3 or 4 percent added silicon is not susceptible.

With regard to hydrogen-stress cracking, there are indications of minor effects of composition on the sensitivity to cracking. However, there is no evidence to indicate any martensitic steel exists which will not break under severe conditions. Variations in steel composition are known to affect the rate at which hydrogen is picked up from the environment and the rate at which it diffuses in the bulk steel. Thus, variations in the time to produce hydrogen-stress cracking do not necessarily reflect differences in the amount of hydrogen required to produce a crack. Unfortunately, it is not yet feasible to separate out the entryrate and diffusion effects to determine whether differences in composition have an effect on the cracking mechanism itself.

In the authors' laboratory, a 5 percent chromium steel (Type H-11) has taken longer to crack under given charging conditions than AISI 4340 steel of the same strength level (260,000 psi ultimate tensile strength) but of lower chromium content (0.85 percent), and longer to crack than AISI 4130 steel (0.87 percent chromium) at a considerably lower strength level (180,000 psi ultimate tensile strength). Cotton⁽²⁰⁾ studied the hydrogen embrittlement of five aircraft-quality high-strength steels plated in three cyanide cadmium baths, as well as in other types of baths. The five steels were AMS 6427 (AISI 4330), AMS 6407, AMS 6418, AISI 4340, and 4340-1, 5Si heat treated to strength levels in the range of 220,000 to 300,000 psi, depending on the nature of the steel. The compositions were as follows:

Constituent	AMS 6427 Steel	AMS 6407 Steel	AMS 6418 Steel	AISI 4340 Steel	4340-1, 5Si Steel
Carbon	0.30	0.28	0.28	0.42	0.38
Manganese	0.88	0.74	1.28	0.78	0.46
Silicon	0.30	0.62	1.40	0.21	1.64
Nickel .	1.80	2.08	1.75	2.05	2.05
Chromium	0.86	1.15	-	0.85	0.98
Molybdenum	0.45	0.44	0.38	0.29	0.36
Phosphorus	0.014	0.017	0.024	0.026	0.010
Sulfur	0.014	0.013	0.010	0.011	0.017
Vanadium	0.08	-	-	-	0.06
Aluminum	-	-		-	0.095

Static-loading tests were used, and the criterion for nonembrittlement was the ability of specimens given the same treatment to withstand a load of 75 percent of the notched ultimate tensile strength for 150 hours without failure. After being plated, all of the specimens were baked at 375 or 550 F for various periods ranging from 3 to 23 hours. Cotton listed the steels in order of increasing resistance to hydrogen embrittlement caused by cadmium electroplating, based solely on time to failure, as follows (strength level in parentheses): AISI 4340 (260,000 to 280,000 psi), AMS 6427 (AISI 4330) (220,000 to 240,000 psi), 4340-1.5Si (270,000 to 300,000 psi), AMS 6407 (220,000 to 240,000 psi), AMS 6418 (220,000 to 240,000 psi). At these high strength levels, there was no consistent relationship between susceptibility to damage and strength level. These results showed that factors other than strength level and ductility of the steel are also important in determining the susceptibility of a steel to hydrogen damage. The silicon-modified 4340 steel (1.5 percent silicon) was more resistant to hydrogen-stress cracking than was the standard AISI 4340 steel, even though the strength level of the former was slightly higher. Of particular significance is the finding that the 4340-1.55i steel at the 270,000 to 300,000-psi strength level was more resistant to hydrogenstress cracking than was the AMS 6427 (4330) steel

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at the 220,000 to 240,000-psi strength level. For the three steels with 0.28 to 0.30 percent carbon used at the 220,000 to 240,000-psi strength level, the high-silicon AMS 6418 steel (Hy-Tuf; 1.40 percent silicon) was more resistant than were the steels of lower silicon content. Thus, it appears that silicon imparts a certain measure of resistance to hydrogen-stress cracking. Some investigators attribute this increased resistance to hydrogen-stress cracking of high-strength steels that results from silicon additions to the higher tempering temperatures used to achieve a given strength level in the silicon steels which produces a different fine structure of the tempered martensite as compared with that in the corresponding low-silicon steel. However, other investigators have shown that silicon and chromium additions reduce the diffusivity or the permeability of hydrogen in α -iron. (21-24) Thus, one would expect the induction period before the onset of hydrogen-stress cracking under sustained load to be appreciably longer in high-silicon and highchromium steels.

Generally, various high-strength steels of a given strength level seem to exhibit about the same minimum stress for failure by hydrogenstress cracking (these lower stresses involve longer times where entry-rate effects are overridden). At high strength levels, differences in notch sensitivity complicate the situation; thus, a difference in breaking time between different materials at a high applied stress does not necessarily mean that there is a difference in sensitivity to hydrogen-stress cracking. For lower strength materials, variations in composition seemed to have little effect on the minimum hardness for cracking under severe cathodic charging conditions, as shown by Table 3.

The available data do not provide sufficient information to determine whether, for a given hydrogen content in the steel, there is any difference in hydrogen-stress-cracking behavior as a result of differences in composition.

In summary, there is not enough information available to determine whether the behavior in regard to differences in composition would tend to suggest that hydrogen-stress cracking and stresscorrosion cracking are the same phenomenon or distinctly different phenomena. More work, performed under more carefully controlled conditions than have been used in the past, is needed to ascertain the effects of composition on these two fracture mechanisms, particularly for hydrogen-stress cracking.

Structure

Whereas steels of all types (including martensitic, bainitic, pearlitic-ferritic, and austenitic) are subject to stress-corrosion cracking, hydrogenstress cracking is common only in steels with martensitic or bainitic structures. For many years it

TABLE 3. MINIMUM HARDNESS IN TUBING STEEL AT WHICH CRACKING OCCURRED(a)(2)

Steel	Minimum Hardness for Cracking, Rockwell C
N-80	17
3-1/2 percent nickel	20
5 percent nickel	20
9 percent nickel	17
9 percent chromium	21
12 percent chromium	19
AISI 4140	20

(a) U-bend test specimens were exposed to artificial hydrogen-embrittlement tests. Cathodic charging in 5 percent H₂SO₄ containing 20 mg/ liter of arsenic as poisoner at 2 amp/in.² for 105 minutes.

was thought that austenitic steels were peculiarly immune to delayed failure induced by hydrogen and that only steels with body-centered cubic structures were susceptible. Cathodic charging of bulk specimens of austenitic steel, even under severe conditions for extended periods of time, resulted only in a partial loss of tensile ductility, but no delayed cracking. However, recently, hydrogen-stress cracking of a thin film of austenitic stainless steel was achieved in the laboratory. It appears as though it is extremely difficult to obtain hydrogen penetration beyond a thin surface layer of a bulk sample of austenite by cathodic charging at room temperature, even though austenite has a high solubility for hydrogen as compared with ferrite. It has been established that the diffusivity of hydrogen in austenite is low compared with that in ferrite of the same composition, and the presence of large amounts of chromium reduces it further (at least in ferrite).

A series of stress-corrosion experiments was performed on a stainless steel that can be treated to have either a martensitic or an austenitic structure. ⁽²⁵⁾ First, the steel was refrigerated to form a martensitic structure. Evaluation showed that chemical attack was randomly nucleated and the structure was resistant to stress-corrosion cracking. When the austenitic version of the same alloy was deformed, coarse slip steps were formed, and these proved to be chemically active sites; this structure was susceptible to transgranular stress-corrosion cracking.

Apparently, the stress-corrosion cracking of austenitic stainless steel occurs only rarely at room temperature, but commonly at elevated temperatures. (6)

As is shown in Figures 1 and 2, hydrogenstress cracking has been obtained in steels with structures consisting of a mixture of pearlite and ferrite, but only rarely has this been observed. Most steels with this structure apparently are not strong enough to crack by this mechanism, since well-authenticated cracks of this type have been obtained only for tensile strengths of 100,000 psi and above (yield strength of approximately 80,000 psi and above). It would appear that the bulk of the pearlitic-ferritic steels can undergo their normal mode of ductile failure at stresses below that required for the hydrogen-stress-cracking mechanism to become operative.

Figures 1 and 2 also show that a steel treated to have either a bainite or a temperedmartensite structure at the same strength level (190,000-psi nominal ultimate tensile strength) exhibited essentially the same hydrogen-stresscracking behavior in the two conditions. Other data in these figures represent the behavior of structures ranging from pearlite with a little ferrite to martensite tempered at a low temperature; all structures were susceptible to hydrogenstress cracking. The times to failure and the minimum stresses for failure correlated very well with the ultimate-tensile-strength level.

The maraging steels, in which hardening is achieved by precipitation hardening of a lowcarbon iron-nickel martensite, are susceptible to hydrogen-stress cracking, but they exhibit reduced sensitivity as compared with conventional martensitic steels. (26,27) There is evidence that the reduced sensitivity observed for these steels is the result of a lower hydrogen-entry rate than is found for conventional martensitic steels. (27)

Very recently, Uhlig and coworkers⁽²⁸⁾ cold worked an austenitic stainless steel, thereby introducing much ferrite into the structure, and succeeded in cracking the mixed structure completely through the section upon charging it with hydrogen. Apparently their test specimens were bent beyond the elastic limit. The strength level of the highly alloyed ferrite in this mixed structure was unknown.

Other investigations have shown that gross variations in structure (including spheroidite, coarse and fine lamellar pearlite, and tempered martensite) have considerable influence on the rate at which hydrogen enters steel and on its permeation and diffusion through steel. Apparently, there is no clear-cut evidence that variations in structure have any influence on the amount of hydrogen required to cause cracking at a given strength level, since a method for determining the amount of hydrogen at the advancing crack tip has not yet been developed. Of course, variations in the entry and diffusion of hydrogen would affect the time required to reach a critical concentration of hydrogen at a site suitable for initiation of cracking.

An investigation was performed to ascertain the minimum applied stress to cause cracking in Type 410 stainless steel heat treated so as to have different amounts of martensite and tested as bentbeam specimens in aqueous hydrogen sulfide solution. The hardness increased with increased percentage of martensite in the structure (see Figure 5). As shown in Figures 6 and 7, above 75 percent martensite (Rockwell C 48 or 140,000-psi yield strength), the steel broke by sulfide cracking at all applied stresses. At the other extreme, no failures occurred in structures that contained less than 32 percent martensite (Rockwell C 30 or 62,000-psi yield strength), not even with applied stresses as high as 97 percent of the yield strength.







FIGURE 6. CURVE SHOWING SULFIDE-CRACKING BEHAVIOR AS RELATED TO % MARTEN-SITE AND APPLIED STRESS (AS POUNDS PER SQUARE INCH)⁽³⁾



FIGURE 7. CURVE SHOWING SULFIDE-CRACKING BEHAVIOR AS RELATED TO PERCENT MARTENSITE AND APPLIED STRESS (AS A PERCENTAGE OF THE YIELD STRESS)⁽³⁾

It is reported that a specific structure is required to make the new French steel (referred to previously) resistant to sulfide cracking. It is claimed that the safe structure consists of ferrite without carbon in solid solution and which contains finely dispersed carbides (cementites partially substituted with chromium or molybdenum and carbides of molybdenum and vanadium). This structure is obtained by water quenching and tempering. Quenched-and-tempered steels can have either acicular or nonacicular structures, and Herzog $^{(29)}$ found that the nonacicular structures performed better in bend tests conducted in sulfide solutions. At the same levels of yield strength or ultimate tensile strength, specimens with the acicular structure broke in 400 to 700 hours, but those with the nonacicular structure withstood more than 2000 hours' exposure under stress.

As was discussed previously, there is no general agreement as to whether sulfide cracking is stress-corrosion cracking, hydrogen-stress cracking, or sometimes both.

Other information pertaining to the effect of structure was obtained in studies of Type 410 stainless steel exposed to pressurized water at 300 F with no imposed potential. ⁽⁵⁾ All samples from two heats had failed in 2 weeks or less under stress. None of the samples from the third heat had failed after 8 weeks' exposure. The structure of the two heats consisted of tempered martensite, and the cracks that developed upon exposure were typical intergranular cracks. However, the structure of the heat that was resistant to cracking consisted of tempered martensite plus longitudinal stringers of delta ferrite; also, the prior austenite grain size of this material was smaller. After 8 weeks' exposure at an applied stress of 40,000 psi, small microcracks were present in the martensite, but propagation of these cracks had been stopped by the delta-ferrite stringers. For exposure at

80,000-psi applied stress, the presence of the delta ferrite appeared to have slowed the crack propagation. The slowing action apparently resulted whenever the crack had to deviate from a direction normal to the direction of the applied stress. Evidence was obtained which suggested that delta ferrite will not stop crack propagation whenever the crack moves in a direction parallel to the stringers.

There is a question, however, whether this phenomenon of cracking in water is stress-corrosion cracking or hydrogen-stress cracking. Had the sensitivity to cracking in the experiments just described also been observed with very small cathodic potentials impressed on the system to make it just a little more cathodic, the results would have been most revealing. As will be discussed in the section dealing with environment, a decreased sensitivity under these conditions would have indicated stresscorrosion cracking as the mechanism acting under the conditions originally used. If the mechanism were hydrogen-stress cracking, cracks would not have been expected to cross the ferrite in this material in which hardness measurements indicate that delta ferrite is considerably softer than the martensite. However, in some precipitationhardened stainless steels, the delta ferrite is age hardened markedly, and a ferrite of this nature might not retard either stress-corrosion or hydrogen-stress cracks.

Sufficient ferrite in a matrix of austenitic Type 304 stainless steel, which strengthens the austenite, reportedly stops stress-corrosion cracks initiated in the austenite.

A fine-grained steel has been reported to exhibit greater resistance to stress-corrosion cracking in caustic than does a steel of coarser grain size of the same carbon content. (18) However, it is not clear whether this is a structural effect or is related to the aluminum probably used to obtain the fine grain size.

Schwier⁽³⁰⁾ performed stress-corrosioncracking experiments on portions of a given lot of steel which had been subjected to various heat treatments. Specimens were stressed in nitrate solution at up to 82 percent of their ultimate tensile strength. The best resistance to cracking (no cracking in 7000 minutes) was exhibited by drawn material (85,000-psi yield strength) and drawn-plus-stress-relieved material (108,000-psi yield strength); air-patented material (43,000-psi yield strength) was intermediate in resistance; while material in the oil-hardened condition (82,000-psi yield strength) exhibited the least resistance (failed in less than 160 minutes).

In summary, there is limited information on cracking in caustic or nitrate solutions which indicates that structure has an effect on the stresscorrosion phenomenon. Other data, which pertain to sulfide cracking and the cracking of Type 410 stainless steel in hot, pressurized water, belong to a gray area in which the mechanism of cracking is uncertain. On the basis of limited information,

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gross differences in structure (for example. tempered martensite versus tempered bainite) appear to have very little effect on the hydrogenstress cracking of high-strength steels at a given strength level, although variations in structure are known to affect hydrogen entry into, and movement in, steel. Of course, variations in the fine structure, such as those that result from tempering martensite at different temperatures, have an effect on sensitivity to hydrogen-stress cracking, the sensitivity being directly related to the strength level, as was discussed in the section dealing with strength level. It might be expected that changes in the fine structure also would affect the entry and movement of hydrogen through steel. Another possible difference in the two modes of failure is the behavior of austenitic stainless steel which, of course, is not a high-strength steel. These steels are susceptible to stresscorrosion cracking in suitable environments, but, in the bulk form, are highly resistant to hydrogenstress cracking. These effects of structure constitute additional evidence that the two types of failure are indeed separate and distinct phenomena.

Applied and Residual Stresses

Hydrogen-induced delayed brittle failures may occur over a wide range of applied tensile stress, and for a given specimen configuration, charging conditions, and test procedures, there is a minimum critical value of applied stress below which failure apparently does not occur in a given material. In general, the time for failure is not affected greatly by variations in the applied stress at high levels of stress. However, small variations in stress become significant as the lower critical stress level is approached.

The results of tests of hydrogen-charged material under static load usually are plotted as sustained load versus time to failure. Such a plot for notched steel specimens precharged with hydrogen is shown in Figure 8. Figure 9 shows a series of such plots for samples of the same steel heat treated to three different strength levels. These data show clearly the significant decrease in ability to support a sustained load as a result of hydrogen in the steel. Applied stresses of as little as 40 percent of the yield strength (for a steel of 270,000-psi ultimate tensile strength) caused failure in only a few hours under sustained load. As is indicated in Figure 9, identical notched specimens in the uncharged condition were stressed at high loads and remained unbroken after times of over 250 hours. For a given strength level, there appeared to be only a slight dependence of failure time on the applied stress. Also, the time to failure was of the same order of magnitude, regardless of strength level. Since the mobility of hydrogen may be expected to remain approximately independent of the strength level and applied stress, even in this early work it appeared that the time to failure may be associated with the diffusion of hydrogen; subsequent work established that this viewpoint was correct. These figures clearly show the



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Aged 5 minutes. Case Institute of Technology Charging Condition A, as given in Figure 9.



FIGURE 9. DELAYED-FAILURE TESTS ON SAE-AISI 4340 STEEL HEAT TREATED TO SEVERAL STRENGTH LEVELS, SHOW-ING THE EFFECT OF HYDROGEN ⁽³²⁾

Fixed charging conditions, aged 5 minutes, sharp-notch specimens.

Case Institute of Technology Charging Condition A:

Electrolyte:	4 percent H ₂ SO ₄ in water
Poison:	None
Current density:	20 ma/in. ²
Charging time:	5 minutes
Aging time:	Measured from end of charging
	to start of test.

occurrence of a minimum critical value of applied stress below which delayed failure did not occur in material of a given strength level. For the specimen geometry, notch acuity, charging conditions, and test conditions used in these experiments, the lower critical stress was nearly the same for the three strength levels. However, this is not always the case for other conditions. This behavior seems to be associated with precharged specimens in which hydrogen is continually being lost from the specimens during the period of static loading. Apparently, after only a few hours (less than 10), sufficient hydrogen is lost from the specimens so that delayed failures do not occur. For these precharged specimens, the lower critical stress was raised markedly as the notch acuity was decreased.

The effect of variation in applied stress when smooth (unnotched) specimens were cathodically charged with hydrogen continuously while under static load is illustrated by the data in Figures 1 and 2. Variations in applied stress affected the time to rupture in a similar manner under a wide range of strength levels, compositions, structures and, to a certain extent, hydrogen contents. When other conditions were held constant, there were two ranges of stress which produced different effects on the time to rupture, as is shown in Figure 1. In the higher range of applied stress, the time to rupture was relatively short and was only moderately affected by a change of stress. In the lower range of stress, a slight decrease of stress resulted in a large increase in the time to rupture, sometimes even a hundredfold. In these experiments where the specimens were continuously cathodically charged, delayed failures were obtained over a wide range of applied stresses; failures occurred at stresses as low as 5 percent of the ultimate tensile strength of the uncharged steel. In these and other experiments, these same investigators showed that the minimum applied stress necessary to cause failure is related to the hydrogen content of the steel; it decreased as the hydrogen content increased. Both the minimum stress for failure and the time required to produce failure were relatively unaffected by differences in the composition or gross structure of the steel, but both decreased as the nominal tensile strength of the steel was increased.

The effect of applied stress in the sulfide cracking of high-strength steel is similar to that just described for hydrogen-stress cracking. This is shown by the work of Warren and Beckman, (4) for example, in which the sulfide cracking of highstrength AISI 4140 bolting material was studied. They found that, in the absence of plastic deformation, no bolt failures occurred at hardness levels below Rockwell C 27, not even under stresses approaching the yield strength of the material. In the hardness range of Rockwell C 27 to 55, the cracking susceptibility increased with increases in the hardness (Figure 4) and applied stress (Figure 10). For each hardness, there appeared to be a minimum or threshold stress required to produce cracking. Increasing hardness lowered the minimum stresses

for failure (Figure 11), the following data being typical of those obtained:

Hardness, Rockwell C	Approximate Minimum Stress for Failure, psi
30	49,000
35	22,000
40	12,000

In addition, plastic deformation increased susceptibility to sulfide cracking by lowering the maximum hardness for absence of cracking to Rockwell C 22. This effect is attributed partly to the residual stresses in cold-worked metal.

McGlasson and Greathouse⁽²⁾ also studied the sulfide cracking of heat-treated steels and likewise found that the limiting hardness level below which failures were not obtained depended upon the applied stress. The greater the applied stress, the lower was the critical hardness level. These results are shown in Figure 3.

Schwier⁽³⁰⁾ studied the stress-corrosion cracking of high-strength steel in nitrate solutions. When he plotted his data as applied stress (as percent of ultimate tensile strength) versus time to failure, a curve resembling that in Figure 8 was obtained. That is, increasing the applied stress increased the sensitivity to cracking by reducing the time to failure. Although relatively few data are available which show the effect of applied stress on nitrate or caustic cracking (stresscorrosion cracking), as a general rule, as the applied stress is increased the sensitivity to stresscorrosion cracking is increased.

In welded or heat-treated parts, particularly those of complex shape, residual stresses can be very high, often approaching the yield strength of the material. In both hydrogen-stress cracking and stress-corrosion cracking, the effective applied stress is the sum of the residual internal stresses and the external applied stress of like sign. Highstrength steels can tolerate but small tensile stresses under conditions which favor stress corrosion or hydrogen entry, and the presence of notches merely makes matters worse. Thus, welding and heat-treating procedures and subsequent stress-relieving treatments that will provide a minimum of residual stresses are a necessary part of many fabricating procedures. Cold-worked metal may be susceptible to stress-corrosion cracking, so suitable annealing operations must follow many steps in fabrication.

Many data are available on the role of applied stress in hydrogen-stress cracking, but, except for a few isolated cases, only a few data related to this effect are available for known cases of stresscorrosion cracking (neglecting the controversial case of sulfide cracking). Those data available indicate that applied stress apparently has the same role in both mechanisms of fracture.



FIGURE 10. EFFECT OF APPLIED STRESS ON TIME TO FAILURE OF AISI 4140 STEEL BOLTS EXPOSED IN AN H₂S-H₂O SYSTEM AT 40 C (104 F) AND 250 PSI⁽⁴⁾



FIGURE 11. MINIMUM APPLIED STRESS RE-QUIRED TO PRODUCE CRACKING OF AISI 4140 STEEL BOLTS IN AN H₂S-H₂O SYSTEM AT 40 C (104 F) AND 250 PSI⁽⁴⁾

Environment

From the data presented thus far in this report, it should be evident that hydrogen-induced delayed brittle failures occur when a critical combination of stress, hydrogen content, and time is exceeded, provided conditions permit the hydrogen in the steel to move freely, either under a hydrogenconcentration gradient or under a stress gradient.

Many investigators have studied the effects of variations in hydrogen content. Because the problems in hydrogen analysis are very great as a result of the small amount usually present and the great mobility of hydrogen even at room temperature, many of the investigations have relied upon such criteria as variations in cathodic charging 'time, in current density, in aging time after charging, or in the concentration of an acid pickling solution or time of exposure to acid chemical attack as the basis for evaluating the effects of variations in hydrogen content. Although hydrogen has been introduced into steel under a multitude of conditions, the results of the various investigations nearly all concur in showing that the delayed failures depend directly on the hydrogen content. If the hydrogen can be kept out or can be removed from the steel before the part is subjected to conditions which result in permanent damage, the problem is circumvented. However, this is not readily achieved because of the numerous processing operations that are potential sources of hydrogen and because of the very small amount of hydrogen (reportedly even less than 1 ppm) that can induce failure.

Hydrogen can be introduced into steel in many ways, including cathodic charging, the action of cathodic protection systems designed to minimize corrosion of metal structures, a wide variety of corrosion reactions, acid pickling, a number of cleaning procedures (including cathodic cleaning), electroplating with a wide variety of metals, exposure to high-pressure hydrogen at room temperature or exposure to atmospheres containing even much smaller partial pressures of hydrogen at elevated temperatures (including welding), exposure to water (corrosion) at room temperature or to moist atmospheres during heat treatment, vapor blasting, glow discharge, hydrocarbon lubricants, and hydrocarbon solvents. Insofar as hydrogenstress-cracking behavior is concerned, the nature of the environment seems to be relatively unimportant so long as a certain amount of hydrogen enters the steel. The minimum stress for failure is relatively unaffected by variations in environment, but the time to failure may vary greatly as it reflects variations in entry rate of the hydrogen.

Only nascent, or atomic, hydrogen can enter the steel lattice and diffuse through it; ordinary molecular hydrogen is not soluble in steel. When 14

atomic hydrogen is generated at a steel surface, a portion of it may enter the steel while the remainder combines into the molecular form and passes harmlessly off insofar as the problem of hydrogenstress cracking is concerned. However, various substances affect the rate at which this recombination reaction takes place. Certain substances with high hydrogen overvoltages, known as cathodic poisons, slow down the recombination reaction. thereby giving the hydrogen atoms a greater lifetime and enhancing their chances of entering the steel. Phosphorus, arsenic, antimony, sulfur, selenium, tellurium, and apparently the cyanide (CN⁻) ion are among the cathodic poisons which promote the entry of hydrogen into iron or steel during pickling or cathodic charging. Hydrogen sulfide is a potent cathodic poison. Other compounds, viz., corrosion inhibitors of the type which become strongly chemisorbed on the surface of the polarized cathode at suitable potential values, act to greatly reduce penetration of hydrogen into an iron or steel cathode. The proper use of such inhibitors can reduce the pickup of hydrogen during acid pickling.

The condition of the steel surface, including such factors as surface roughness, the presence of a layer of cold-worked metal, and the presence of oxide films, also influences the entry of hydrogen into steel.

These environmental factors are important in that a greater or lesser amount of atomic hydrogen is presented to the steel. The condition of the surface, as well as the presence of cathodic poisons or inhibitors, affects entry rate. However, once a certain amount of hydrogen has been introduced into a susceptible steel which is at a sustained stress above the lower critical value, the hydrogenstress-cracking mechanism can become operative. It makes little difference just how the hydrogen was introduced into the steel.

If the entry surface and exit surface are in sufficiently close proximity, the hydrogen concentration may never build up to a level high enough to be troublesome. For example, such a condition might exist with sheet material or thin tubing on which atomic hydrogen is generated on one surface, depending on the strength level and sensitivity of the material and on the stress to which it is subjected.

Environmental conditions also have a great effect on the rate at which hydrogen leaves a steel part at surfaces which are not entry surfaces. An increase in temperature is very effective in accelerating the removal of hydrogen. Thus, baking at 400 F removes hydrogen some 250 to 500 times faster than aging at room temperature. However, aging in hot water is considerably more effective than aging in dry air at the same temperature, and other atmospheres also influence the rate of hydrogen removal. Oxide coatings or electroplated metal coatings on the steel surface tend to act as a barrier to the removal of hydrogen. A dense cadmium

electroplate, frequently used for corrosion protection of high-strength steel parts, is virtually impervious to the passage of hydrogen. Thus, hydrogen introduced into steel by a prior heattreating, pickling, or cleaning operation tends to be trapped in the part by the cadmium electroplate. With a dense cadmium electroplate, baking apparently serves merely to redistribute the hydrogen within the part but not to remove it. With porous electroplates, baking may serve to remove much of the trapped hydrogen, but the protective value of the plate against corrosion may be inferior to that of a dense plate. Also, because cadmium is a sacrificial coating and, therefore, is anodic with respect to the steel, hydrogen can be generated cathodically at breaks in the coating where the steel is exposed.

The role played by corrosion appears to be essential to the stress-corrosion-cracking mechanism. When the proper conditions of stressinduced electrochemical corrosion are met in a susceptible material, it is generally agreed that metal goes into solution at the tip of the advancing crack, which is, therefore, an anode. Simultaneously, electrons are given off into the metal through which they flow to a localized region of the metal surface also in contact with the electrolyte. At this second area, the electrons participate in a reduction reaction; hence, the area is a cathode. The cathode reaction may be the reduction of positively charged hydrogen ions to atoms, which may in turn react with dissolved oxygen to form water, may recombine into molecular hydrogen and bubble off as hydrogen gas, or may dissolve in the metal. The last-mentioned possibility may give rise to simultaneous hydrogen-stress cracking in highstrength steels. However, some cathode reactions do not involve the reduction of hydrogen ions, and atomic hydrogen is not liberated at these local

cathodes. The thermodynamics of the reactions of iron indicate that hydrogen is liberated in acid solutions (below about pH = 4) and in highly alkaline solutions (above about pH = 10). In nitrate solutions, nitrites usually form at the cathode, and in weakly alkaline solutions, peroxides generally are formed. Although stress-corrosion cracking occurs at a local anode, the cathodic reaction must be maintained by removal of the corrosion products, for if the cathode reaction stops (cathode becomes polarized), the anode reaction likewise stops. The cathode area may be outside the crack or deep within it.

The propagation of the stress-corrosion crack depends at least partly upon the flow of current between the anodic and cathodic areas and, thus, is electrochemical in nature, as was mentioned above. If the electrochemical potential of the cathodic areas were adjusted to match the potential of the anodic areas, there no longer would be a driving force for the electrochemical reaction, no current would flow, and crack propagation would be expected to stop. Such an adjustment of the potential can be made by impressing small direct

currents from an external source, and the procedure is called cathodic protection. This arrangement has been demonstrated to be effective in stopping or at least retarding cracking in several stress-corrosion systems. (6, 33)

Brown^(34,6) and Bhatt and Phelps⁽³⁵⁾ have proposed a procedure to be used in attempting to distinguish stress-corrosion cracking from hydrogen-stress cracking. It is based on the assumptions that (1) the term "stress-corrosion cracking" will be applied to the cracking that involves corrosion which occurs at anodic areas at the advancing tip of the crack, (2) electric currents flow between local anodes and local cathodes to effect this corrosion, and (3) impressing small cathodic currents from an auxiliary electrode should tend to polarize the local cathode areas in the direction of the potential of the anodes and thereby should tend to mitigate the corrosion and the attendant cracking. Figure 12 shows the effect of small currents (note that these are of the order of microamperes per square centimeter) on the



Impressed Current, microamperes per cm²

FIGURE 12. EFFECT OF IMPRESSED CURRENTS ON TIME TO FRACTURE OF A MAR-TENSITIC STEEL IN HYDROGEN SULFIDE SOLUTION(6)

> A positive current in Figures 12, 13, and 14 indicates that the specimen is being made cathodic; a negative current indicates anodic.



FIGURE 13. EFFECT OF IMPRESSED CURRENTS ON TIME TO FRACTURE OF A MAR-TENSITIC STEEL IN ACETIC ACID WITH SULFIDE POISON(6)

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time to fracture of Type 410 stainless steel in a hydrogen sulfide solution. The fact that a low level of impressed cathodic current very greatly extended the breaking time was assumed to indicate that a corrosion reaction was stifled and the cracking process in this environment in the absence of impressed current is properly classified as stresscorrosion cracking. Note that at high current densities, the specimens also fractured after a short time; this was assumed to be hydrogenstress cracking from the hydrogen that was evolved cathodically.

Figure 13 shows the results of a similar study performed in a different medium. Here, imposing even a small cathodic current (evolution of hydrogen) decreased the breaking time, whereas the imposition of anodic currents (oxidizing conditions and reduction of hydrogen) greatly extended the life of the specimens. This behavior was believed to indicate that, in the case of no impressed current (that is, the "natural" cracking case), the failure occurs by the hydrogen-stress-cracking mechanism.

The results of a similar study performed with a solution of NaCl that contained acetic acid and hydrogen sulfide, a particularly agressive cracking medium, are shown in Figure 14. In this situation, impressing either anodic or cathodic currents on the specimen had no effect on the breaking time. Brown suggested that one possible explanation for this behavior is that both stress-corrosion cracking and hydrogen-stress cracking can occur in overlapping ranges of current density.

Brown⁽⁶⁾ cautioned that these electrochemical analyses must be viewed as highly simplified and preliminary. However, he reported that they were consistent with two subsequent observations: (1) conditions presumed by this analysis to induce stress-corrosion cracking caused the measured potential to behave during cracking as if the cracking were occurring along an active path, and (2) conditions presumed to induce hydrogen-stress cracking caused hydrogen to permeate a thin iron membrane.



FIGURE 14. EFFECT OF IMPRESSED CURRENTS ON TIME TO FRACTURE OF A MAR-TENSITIC STEEL IN MIXTURE OF NaCl, ACETIC ACID, AND SUL-FIDE⁽⁶⁾

Scharfstein and Eisenbrown have detected differences in the potential-time curves obtained during stress-corrosion cracking and hydrogenstress cracking.⁽³⁶⁾ Typical potential-time curves for stress-corrosion cracking were described as follows:

- There was a gradual rise in potential in the noble direction immediately after the specimen was exposed to the cracking medium.
- (2) After a limiting potential was reached, there were rapid drops in potential to a more active value, followed by a rapid return to the normal value.
- (3) When the specimen fractured, there was a very rapid drop to a more active potential.

Upon exposing a 12 percent chromium martensitic stainless steel to 1%HCl-1%SeO₂ (a demonstrated promoter of hydrogen-stress cracking), the following potential values were obtained:

- There was a gradual drop to a more active potential during the first 30 minutes.
- (2) After the lowest potential was reached, there were rapid rises in potential followed by a rapid return to normal.
- (3) When the specimen failed, there was a sudden rise to a more noble potential.

When the 12 percent chromium steel was exposed to 5%H2SO4-0.1%(NH4)2S (promotes hydrogen-stress cracking), the potential dropped to a more active value and remained there until cracking occurred. Thus, the potential-time curves were exactly opposite for stress-corrosion cracking and hydrogen-stress cracking. The difference in the shape of the potential curves for the two hydrogen-cracking media was attributed to reduction of selenious acid and deposition of selenium metal in the HCl solution. This produced a steady potential until fracture occurred. Fresh metal was then exposed, and the potential rose. In the sulfuric acid solution, only hydrogen was reduced and there was no competing process, thus accounting for the lowering of the potential to a steady value.

In general, stress-corrosion cracking requires a highly anodic area and localized low pH to get started. Examples of such sites are cracks in an oxide film, pits, crevices, and cold-worked material, the latter being highly anodic.

Some have raised the question of whether in stress-corrosion cracking there is an actual loss of metal or the material is merely being charged with hydrogen. In many environments very little corrosion takes place, and frequently it cannot be detected with the naked eye. However, in other instances, loss of metal and corrosion products have been observed. With the electron microscope, Swann, Nielsen, and others have detected corrosion tunnels, often nucleated in rows on the surface of an alloy. Oxide has been detected in the stresscorrosion cracks formed in austenitic stainless steel in certain environments. However, stresscorrosion cracking in MgCl₂ does not appear to be associated with a solid corrosion product. NaCl is an aggressive corrodent; although it enters into the corrosion reaction, it is not found in the corrosion product. Its action is to hinder formation or repair of protective surface films.

Although a highly corrosive environment will not produce stress-corrosion cracking of highstrength steel, some such environments do result in hydrogen-stress cracking, for example in the acid pickling of steel.

Anhydrous agricultural ammonia will crack steel tanks at room temperature, but when a certain amount of water is added, the cracking is elim-Since the reaction might involve disinated. sociated hydrogen, this may be hydrogen-stress cracking rather than stress-corrosion cracking. This type of cracking has been observed in steel plate with yield strengths of 37,000, 51,000, 48,000, and 57,000 psi, for example, all values which would appear to be too low for hydrogenstress cracking. However, the cracking has occurred mostly near welds, and the strength of weld heat-affected zones frequently is considerably higher than that of the parent metal. The cracking has been reported to be intergranular in the heataffected zone but transgranular in the parent metal. In unwelded steel, such cracking resulting from exposure to agricultural ammonia has been observed in T-1 steel (116,000-psi yield strength). This steel is also susceptible to hydrogen-stress cracking when cathodically charged with hydrogen. The addition of 0.25 percent water eliminates the cracking in the ammonia environment. This may be related to a critical moisture content which forms a protective barrier film, since in other instances an excess of water aids in introducing hydrogen into high-strength steel. The presence of air increases the cracking rate resulting from exposure to ammonia; this is believed to be the result of CO2 present in the air.

Several instances of cyanide cracking have been reported; these failures are not well documented but structural steel was reported to be involved. Concentrations of only 2.6 to 3.5 g HCN/liter have caused cracking.

Stress-corrosion cracking has been observed in various nitrate solutions, including boiling mixtures of calcium and ammonium nitrates, 60 to 95 percent ammonium nitrate, concentrated nitric acid at 20 C (68 F), and dilute nitric acid that contained small amounts of MnCl₂ used at 60 to 80 C (140 to 175 F). The caustic cracking of steel is well known. Concentrations range from the small amounts used in boiler feedwater to at least 50 percent NaOH used in chemical-processing operations. Numerous failures have occurred in boilers at elevated temperatures and in storage tanks, usually at elevated temperatures but also at ambient temperatures. Agents added to boilers to inhibit cracking include tobacco juice, tannin, nitrates, sulfates, phosphates, waste sulfite liquor, lignin, bark extract, and quebracho extract. Cracking can be accelerated by the addition of silicates, PbO, MnO₄, or Na₂CrO₄.

It is believed that the stress-corrosion cracking of austenitic stainless steels generally is related to chlorides or caustic. However, stress-corrosion cracking of high-strength martensitic steels can result from exposure to any of a large number of different types of environments, even including humid argon. Some very mild media have resulted in stress-corrosion cracking of specimens with coatings generally considered to be good barriers against corrosive attack. Water free from chloride ions will crack Type 410 martensitic stainless steel or 17-4PH stainless steel. but not low-strength steels. However, this may be a case of hydrogen-stress cracking. It is not necessary for the corrodent to be a bulk aqueous solution; it may be a thin adsorbed layer of an aqueous solution.

Cracking is produced in aqueous solutions containing 15.2 ppm of hydrogen sulfide and even less. ⁽²⁾ The term sulfide-corrosion cracking has been used by many to describe the cracking of steel in sulfide environments. There has been no complete agreement among investigators whether it is a stress-induced electrochemical phenomenon with loss of metal occurring at the tip of the advancing crack and, hence, stress-corrosion cracking, or whether it is a hydrogen-stress-cracking mechanism that is operative. Many are inclined to the latter viewpoint. However, some of the data of Brown discussed above suggest that both mechanisms may be operative simultaneously in at least some instances of sulfide cracking.

The temperature of the environment is important in both hydrogen-stress cracking and stress-corrosion cracking. The maximum susceptibility to hydrogen-stress cracking occurs at about room temperature. The decreased susceptibility at lower temperatures results from the decrease in the hydrogen-diffusion rate with decreasing temperature. Since hydrogen does not diffuse so rapidly to the region of triaxial stress just in advance of the tip of the crack, the required hydrogen concentration cannot be built up so rapidly and crack propagation is slowed or stops altogether at sufficiently low temperatures (such as the temperature of liquid nitrogen). Troiano and coworkers at Case Institute of Technology (37) have suggested that the disappearance of hydrogen embrittlement and hydrogen-stress cracking at somewhat elevated temperatures (300 or 400 F)

also may be related to stress-induced hydrogen diffusion. Because the driving force tending to concentrate hydrogen in the region of maximum triaxiality is stress induced, the driving force presumably is independent of temperature. However, the force tending to homogenize the solid solution presumably increases with temperature. Thus, at elevated temperatures, the required hydrogen gradient cannot be created and embrittlement and cracking tendencies should decrease, just as has been observed.

Susceptibility to stress-corrosion cracking increases as the temperature increases, and particularly as the temperature is raised above the normal boiling point of aqueous solutions. However, this behavior is not observed with hydrogen sulfide solutions; for these, the susceptibility is greatest near room temperature, just as for hydrogen-stress cracking.

If notched specimens of high-strength steels are broken over a range of temperatures, they exhibit a transition from brittleness at lower temperatures to toughness at higher temperatures. Because of this transition-temperature effect, the length of crack required to set off catastrophic brittle fracture is increased as the test temperature is increased in the transition range (which for many high-strength steels and a number of specimen geometries is in the general region of room temperature). Thus, if one attempts to measure a temperature coefficient of either hydrogen-stress cracking or stress-corrosion cracking by observing merely the effect of temperature on time to failure, he may be observing the combined effects of change in temperature on true stress-corrosioncracking and hydrogen-stress-cracking rates and on purely mechanical fracture.

In gaseous atmospheres, the entry rate of hydrogen into steel increases as the partial pressure of dissociated hydrogen atoms in the atmosphere increases. Few data are available regarding a pressure effect in stress-corrosion cracking, but this phenomenon generally involves an aqueous environment or molten salts.

To summarize the effects of environment, it has been shown that both stress-corrosion cracking and hydrogen-stress cracking can occur under a great variety of environmental conditions. However, it appears that certain environments favor only one type of cracking or the other. This observation, the observed difference in behavior of the two types of cracking under impressed electrical currents, and the different behavior observed for variations in temperature suggest that hydrogen-stress cracking and stress-corrosion cracking actually are different fracture mechanisms. However, there is an area that includes sulfide cracking and cracking in water where the distinction is not clearcut. It may be that both types of cracking are operative in these systems, possibly sometimes simultaneously.

Time

It is obvious from previous discussions that time is a factor in both mechanisms of cracking. Many factors related to the environment, such as composition, concentration, and temperature, can affect the rate at which atomic hydrogen is liberated at the steel surface or the aggressiveness of the corrosion reaction, and other factors (such as steel composition, the presence of cathodic poisons, and the nature of the steel surface) affect the rate at which hydrogen enters the bulk steel. These various factors, of course, influence the time required to nucleate a crack. Apart from the matter of getting hydrogen into the steel, the hydrogenstress-cracking mechanism itself is diffusion controlled. Hydrogen must diffuse to the region of maximum triaxial stress a short distance in advance of the crack tip before the crack can propagate another incremental distance. As the crack propagates, the region of triaxial stress also moves, so, in most instances, hydrogen must move if crack propagation is to continue. The hydrogen and the crack thus move in stepwise fashion until the crack has grown to the point where the remaining section can no longer support the load, and terminal catastrophic failure ensues. Stress-corrosion cracking proceeds by the timedependent corrosion at the tip of the crack which makes possible a cohesive separation at low stress, and this proceeds until the section suddenly fails mechanically from overstress.

If, under the same charging conditions, two steels differ in hydrogen entry rate, the steel with the lower entry rate requires a longer time to build up the necessary amount of hydrogen in a critical area where a crack can nucleate. If the same amount of hydrogen is required to crack the two steels, then that one with the low entry rate would take longer to crack and would be considered to be less susceptible to hydrogen-stress cracking. There is considerable evidence to support differences of this type among various steels. Among high-strength steels, there is little evidence to date to indicate great differences in the ability of steels to tolerate different amounts of hydrogen.

If, under given test conditions, one steel is less notch sensitive than another, it may require a longer time to fracture by either stress-corrosion cracking or hydrogen-stress cracking. This results not because of a lower rate of stress corrosion or of hydrogen entry but because a longer crack must be produced before it is of a critical size to result in the terminal catastrophic fracture which is purely mechanical in nature. High-strength steels usually are more notch sensitive than are those of lower strength, so it is not necessary for a crack to grow to as great a length in the former before catastrophic terminal failure. This automatically gives shorter failure times as the strength level is increased. Also, for purely geometric reasons, a thick steel specimen is more susceptible to brittle fracture than is a thin one. This applies not only to fractures of mechanical origin but to hydrogen-stress cracking and stress-

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corrosion cracking as well. Thus, a stresscorrosion crack or a hydrogen-stress crack may not have to be nearly so extensive in a thick specimen as in a thin one to set off catastrophic brittle failure.

In the laboratory where it was known that cracking conditions prevailed continuously, the longest delay known to the authors of this report was 355 days. This was the time to visible stresscorrosion cracking of an 18Ni maraging steel (303,000-psi yield strength) exposed to natural seawater. ⁽³⁸⁾ Under these same conditions, AISI 4340 steel (240,000-psi yield strength) cracked in 6 days. In the authors' laboratory, a sample of carbon-manganese steel heat treated to a yield strength of 156,000 psi (169,000-psi ultimate tensile strength) failed after 1147 hours under mild cathodic charging conditions; this is the longest delay before failure by hydrogen-stress cracking of which the present authors are aware.

Both fracture mechanisms are time dependent, and the time variable does not appear to provide a means of distinguishing between them.

Fracture Characteristics

Fractures produced by either stresscorrosion or hydrogen-stress cracking in highstrength steels are characterized by almost complete lack of visual evidence of plastic deformation. There is considerable evidence to indicate that both types of cracks tend to grow in a direction normal to the maximum tensile stress. Furthermore, both types of cracks are predominantly intergranular in character in high-strength steels, although some transgranular cracks also are observed.

In general, hydrogen-stress cracking produces a few nonbranching cracks. Stresscorrosion cracking usually produces several secondary cracks, in addition to the major crack; both the major crack and the secondary cracks often are branching.

In general, the fracture surfaces appear quite similar in the two mechanisms of fracture. However, certain subtle differences have been observed which some investigators have attributed to the different cracking processes. For example, Tiner et al⁽³⁹⁾ working with AISI 4340 steel in aerated 3 percent NaCl solution have separated the fracture into three stages: incubation, stable crack-growth period, and final rupture. During the incubation stage, cracks were fan shaped and intergranular, and they extended to a depth of several mils. In the stable-crack region (greater than 5 to 6 mils below the surface), the crack path in relationship to the prior austenite grains was found to be both intergranular and transgranular. The intergranular portion was related to a stress-corrosion-cracking mechanism, while the transgranular path was attributed to hydrogenstress cracking. These fracture areas were identified in electron-microscope fractographs. The final stage was tensile rupture of the material.

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