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INTERIM SUMMARY OF SANDIA CREEP EXPERIMENTS ON ROCK SALT  
FROM THE WIPP STUDY AREA, SOUTHEASTERN NEW MEXICO

MASTER

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# CONTENTS

|  | Page |
|--|------|
| INTRODUCTION . . . . .                                       | 1    |
| PRIMARY & BASELINE (QUASI-STATIC) EXPERIMENTS . . . . .      | 3    |
| OTHER EXPERIMENTS . . . . .                                  | 7    |
| Apparatus and Experimental Procedures . . . . .              | 7    |
| Instrumentation, Measurements and Data Acquisition . . . . . | 9    |
| MATERIAL DESCRIPTION . . . . .                               | 11   |
| TEST MATRIX . . . . .  | 13   |
| RESULTS . . . . .  | 14   |
| General Features . . . . .                                   | 17   |
| Primary (Transient) Creep . . . . .                          | 18   |
| Secondary (Steady State) Creep . . . . .                     | 20   |
| Tertiary Creep and Creep Fracture . . . . .                  | 23   |
| Effect of Stress or Strain History . . . . .                 | 23   |
| PRELIMINARY COMPARISON OF RESULTS WITH RE/SPEC DATA          |      |
| AND DISCUSSION . . . . .                                     | 24   |
| SUMMARY . . . . .  | 27   |
| ACKNOWLEDGEMENTS . . . . .                                   | 29   |
| REFERENCES . . . . .   | 30   |
| FIGURES . . . . .  | 32   |
| APPENDIX (Condensed Creep Data Files) . . . . .              | 52   |
| DISTRIBUTION . . . . .                                       | 65   |

# TABLES AND FIGURES

|   | Page |
|---|------|
| TABLE 1 . . . . .   | 5    |
| Ultimate stress and strain data of short-term loading tests.  |      |
| TABLE 2 . . . . .   | 13   |
| Creep test matrix.  |      |
| TABLE 3 . . . . .   | 15   |
| Summary statistic of creep tests.   |      |
| TABLE 4 . . . . .   | 19   |
| Summary of primary (transient) creep data.  |      |
| TABLE 5 . . . . .   | 21   |
| Summary of secondary creep estimates.   |      |
| FIG. 1 . . . . .  | 27   |
| Triaxial compression apparatus<br>Parts key: (1) load frame (2) hydraulic actuators<br>(3) pressure vessel (4) loading ram (5) bottom<br>(6) insulation   |      |
| FIG. 2 . . . . .  | 28   |
| Typical variations of principal stress difference with time<br>for creep tests at 500 psi (3.5 MPa) confining pressure<br>(sample 9-2625).  |      |
| FIG. 3 . . . . .  | 29   |
| Plot of principal stress difference and confining pressure<br>versus time for sample 7-2771. Note extreme variations in<br>( $\sigma_1 - \sigma_3$ ) during periods of constant applied ram load<br>(Fig. 1).   |      |
| FIG. 4 . . . . .  | 36   |
| Axial creep record ( $\epsilon_1$ vs. $t$ ) for sample 7-2777. Note<br>strong influence of updates in ( $\sigma_1 - \sigma_3$ ).  |      |
| FIG. 5 . . . . .  | 37   |
| Axial creep record ( $\epsilon_1$ vs. $t$ ) at ( $\sigma_1 - \sigma_3$ ) = 1000 psi<br>(6.9 MPa), $T = 100^\circ\text{C}$ and variable confining pressure. Note<br>abbreviated test code: drillhole no.-core depth in ft/con-<br>fining pressure in ksi/test temperature in $^\circ\text{C}$ (principal<br>stress difference in psi). |      |

|  |    |
|--|----|
| FIG. 6. . . . .  | 34 |
| Radial creep records ( $\epsilon_3$ vs. $t$ ) at $(\sigma_1 - \sigma_3) = 1000$ psi (6.9 MPa), $T = 100^\circ\text{C}$ and <u>variable confining pressure</u> (for test code see Fig. 5).  |    |
| FIG. 7. . . . .  | 39 |
| Axial creep records ( $\epsilon_1$ vs. $t$ ) at $(\sigma_1 - \sigma_3) = 4500$ psi (31.0 MPa), $T = 22^\circ\text{C}$ and <u>variable confining pressure</u> (for test code see Fig. 5).   |    |
| FIG. 8. . . . .  | 40 |
| Radial creep records ( $\epsilon_3$ vs. $t$ ) at $(\sigma_1 - \sigma_3) = 4500$ psi (31.0 MPa), $T = 22^\circ\text{C}$ and <u>variable confining pressure</u> (for test code see Fig. 5).  |    |
| FIG. 9. . . . .  | 41 |
| Axial creep records ( $\epsilon_1$ vs. $t$ ) at $\sigma_3 = 500$ psi (3.5 MPa), $T = 22^\circ\text{C}$ and <u>variable principal stress difference</u> (for test code see Fig. 5).         |    |
| FIG. 10. . . . .   | 42 |
| Axial creep records ( $\epsilon_1$ vs. $t$ ) at $\sigma_3 = 500$ psi (3.5 MPa), $T = 100^\circ\text{C}$ and <u>variable principal stress difference</u> (for test code see Fig. 5).        |    |
| FIG. 11. . . . .   | 43 |
| Axial creep records ( $\epsilon_1$ vs. $t$ ) at $\sigma_3 = 500$ psi (3.5 MPa), $(\sigma_1 - \sigma_3) = 1000$ psi (6.9 MPa) and <u>variable temperature</u> (for test code see Fig. 5).   |    |
| FIG. 12. . . . .   | 44 |
| Axial creep records ( $\epsilon_1$ vs. $t$ ) at $\sigma_3 = 3000$ psi (20.7 MPa), $(\sigma_1 - \sigma_3) = 1000$ psi (6.9 MPa) and <u>variable temperature</u> (for test code see Fig. 5). |    |
| FIG. 13. . . . .   | 45 |
| Axial strain records ( $\epsilon_1$ vs. $t$ ) for sample 9-2625. Creep stage C2 proceeded at $(\sigma_1 - \sigma_3) = 4400$ psi (30.3 MPa).  |    |
| FIG. 14. . . . .   | 46 |
| Axial strain records ( $\epsilon_1$ vs. $t$ ) for sample 9-2625. Creep stage C2 proceeded at $(\sigma_1 - \sigma_3) = 3300$ psi (22.8 MPa).  |    |

|   |    |
|---|----|
| FIG. 15 . . . . .   | 47 |
| Double logarithmic plot $\log(\dot{\epsilon}_1)$ vs. $\log(t)$ of test<br>9-2083/1.5/22 (2950) (for test code see Fig. 5).  |    |
| FIG. 16 . . . . .   | 48 |
| Semi logarithmic plot $\dot{\epsilon}_1$ vs. $\log(t)$ of test 9-2083/1.5/22<br>(2950) (for test code see Fig. 5).  |    |
| FIG. 17 . . . . .   | 49 |
| Secondary creep rates $\dot{\epsilon}_1$ for lower level salt in the space<br>( $\log(\dot{\epsilon}_1)$ , $1/T$ ). Triangular data points are due to RE/SPEC,<br>Inc. (Ref. 11). $\Delta\sigma = (\sigma_1 - \sigma_3)$ . Stresses are given in psi,<br>strain rates in ( $s^{-1}$ ). Parentheses denote upper bounds. |    |
| FIG. 18 . . . . .   | 50 |
| Secondary creep rates $\dot{\epsilon}_1$ in space $\log(\dot{\epsilon}_1)$ , $\log(\sigma_1 - \sigma_3)$<br>for lower level salt. Triangular data points are due to<br>RE/SPEC, Inc. (Ref. 11). Stresses are given in psi, strain<br>rates in ( $s^{-1}$ ). Parentheses denote upper bounds.                            |    |
| FIG. 19 . . . . .   | 51 |
| Secondary creep rates $\dot{\epsilon}_1$ in the space $\log(\dot{\epsilon}_1)$ , $\log(\sigma_1 - \sigma_3)$<br>for upper level salt. Stresses are given in psi, strain<br>rates in ( $s^{-1}$ ). Parentheses denote upper bounds.  |    |

## INTRODUCTION

This report presents an interim summary of triaxial creep experiments which have been performed at Sandia Laboratories on rock salt from the WIPP study area near Carlsbad, New Mexico. The data presented support ongoing efforts to formulate suitable constitutive equations to predict the response of rock salt from southeastern New Mexico to the perturbations of a radioactive waste repository.

The creep experiments of this account are part of an initial test matrix which was defined by Sandia Laboratories and which is being filled by Sandia Laboratories and by RE/SPEC, Inc. This matrix was established to measure the time-dependent behavior of New Mexico rock salt as a function of principal stress difference (isotator stress), confining pressure or mean stress and temperature. The importance of these parameters was substantiated by some published data for rock salt and potash (1-8) under low confining pressure  $\sigma_3 = \sigma_2 \leq 1000$  psi (20.7 MPa) which is likely to be typical for the proposed Waste Isolation Pilot Plant. The relevance of all four parameters was further indicated by site specific results of short-term (quasi-static) tests which preceded or accompanied creep experiments in an earlier phase of the Sandia program (9-12).

The creep experiments which will be described here include 22 tests subjected to principal stress differences between 930 psi (6.4 MPa) and 1490 psi (22.1 MPa) and confining pressures between zero (unconfined) and 3000 psi (20.7 MPa). The test temperatures were 22, 100 and 200°C.

The present results, combined with earlier data of RE/SPEC, Inc., indicate further study is required to completely describe the response of New Mexico rock salt under general, time-varying conditions of stress and temperature. To achieve this goal, tests of longer duration are

needed, including creep rupture experiments. Furthermore, extension tests and possibly, straight-pull tension tests are adviseable. Finally, additional experiments should be performed to treat data scatter statistically, to account for the unknown effects of impurities and grain size of New Mexico rock salt and to consider differences in salt core histories.



#### SUMMARY OF BASELINE (QUASI-STATIC) EXPERIMENTS

The nature and sequence of creep experiments completed to date are influenced directly by observations which were made in short-term, so-called quasi-static experiments (12). The majority of these tests were carried out on salt core from drill holes APC #7 and FADA #2 at a near, low loading rate of  $(\dot{\sigma}_1) \leq d(\sigma_1 - \sigma_3)/dt \leq 40$  (0.41 MPa/min) psi/min. This loading rate was chosen primarily to compare the results for New Mexico rock salt with extensive published data for rock salt (for example 6). Supplementary experiments were also performed at loading rates up to approximately 4,000 psi/min (4.14 GPa/min).

The following results are deemed particularly pertinent (13).

(1) New Mexico rock salt in the laboratory is very non-linear under all loading conditions with an initial elastic limit  $(\sigma_1 - \sigma_3)^0 = 1$ . This behavior appeared to be unaffected by differences in hydrostatic pressure history to 1000 psi (34.5 MPa) prior to deviatoric loading. It is reasonable that the low initial elastic limit is influenced by damage during coring in situ and subsequent core handling. However, it is impossible at present to separate out this effect because the exact stress history of the material before laboratory testing remains unknown.

(2) The elastic properties of New Mexico rock salt can be evaluated accurately only in load/unload/reload cycles provided the imposed loading rate is sufficiently high or the range of stresses, either hydrostatic pressure  $p$  or  $(\sigma_1 - \sigma_3)$ , is well below the previously attained peak value. If the elastic constants are evaluated in this manner, then Young's modulus,  $E$ , and Poisson's ratio,  $\nu$ , fall into the ranges  $1.2 \times 10^6 \leq E \leq 1.3 \times 10^6$  psi ( $29.6 \leq E \leq 36.5$  MPa) and  $0.17 \leq \nu \leq 0.26$ . These data compare very favorably with in situ measurements based on records of  $p$ - and  $s$ -wave velocities and rock densities (13).

(4) Unloading and recovery measurements indicate that practically all immediate rock salt deformation is permanent. Additionally, the effect of deformation of New Mexico rock salt subject to deviatoric loading can be treated as a very small fraction of the total deformation.

(5) So far it has been impossible to separate the permanent deformation of rock salt into time-independent and time-dependent components even at the relatively high loading rate,  $d(\sigma_1 - \sigma_3)/dt \approx 20,000$  psi/min (1.4 kN/m<sup>2</sup>/min).

(6) At low confining pressures  $\sigma_3 = \sigma_2 \leq 3000$  psi (20.7 Mpa) New Mexico rock salt exhibits behavior in which deformation approached or became unstable at high strains and in which deformation is associated with large dilatancy due to grain rearranging. It appears that the rock is polycrystalline, possibly because of grain sizes at low confining pressures,  $(\sigma_1 - \sigma_3) \leq 1100$  psi (7.6 Mpa) were relatively small, the elevated confining pressure,  $\sigma_3 = \sigma_2 \approx 3000$  psi (20.7 Mpa) and/or at high temperatures  $T \approx 300^\circ\text{K}$ .

(7) The grain sizes and stability of New Mexico rock salt depend on deformation rate and pressure. According to Table 1, at ambient temperatures the ultimate (true) stresses and the corresponding greatest compressive strains,  $\epsilon_3$ , were then approximately 3500 psi (24.1 Mpa) and 27% in uniaxial compression to 1000 psi (6.9 Mpa) and  $1.1 \leq \epsilon_3 \leq 14\%$  at 500 psi (3.4 Mpa) confining pressure. At 1000 psi (6.9 Mpa) confining pressure no signs of grain collapse were observed at low 800 psi (5.5 Mpa) and  $\epsilon_3 = 2\%$ . Thermodynamic behavior was associated with large anisotropic strain ratios  $\epsilon_2/\epsilon_1$ . For example, when the ultimate stress was reached the ratio  $\epsilon_2/\epsilon_1$  amounted to -1.7, and -0.65 at zero and 500 psi (3.4 Mpa) confining pressure, respectively. By comparison at 3000 psi (20.7 Mpa) confining pressure, the greatest observed principal strain ratio was  $\epsilon_3/\epsilon_1 = -0.69$ . Temperature produced a

# TABLE 1

Table 1: Stress and Strain (Ductility) Data of Short-Term Loading  
 Tests: (1)  $\dot{\epsilon} = 10^{-3}$  in./in. min. (2)  $\dot{\epsilon} = 10^{-1}$  in./in. min.  
 Normal and strain represent true stress and engineering strain, respectively.

| Test Code<br>(Hole# - Depth (ft.)<br>conf. press. (ksi)/Temp. °C) | Ultimate (Max)<br>Engng. Stress Diff.<br>(psi) | Ultimate (Max)<br>Strain (%) | Strain $\epsilon_1$<br>at Ultimate<br>Stress (%) |
|---|--|------------------------------|--|
| Upper Level Salt  |  |                              |  |
| 9-2003/0/23   | 2100   | 14.7                         | 1.6  |
| 9-2003.5/5/23   | 2100   | 14.1                         | 10.5   |
| 9-2001.5/3.7/23   | >2000*   | >2.1                         | >20  |
| Lower Level Salt  |  |                              |  |
| 9-2006/0/23   | 2300   | 10.7                         | 2.6  |
| 9-2006.7/0/23   | 2300   | 10.1                         | 2.4  |
| 9-2009/0.1/23   | 2300   | 9.4                          | 4.6  |
| 7-2007/0/23   | 2200   | 10.7                         | 4.7  |
| 9-2007.5/5/23   | 2300   | 10.9                         | 1.6  |
| 9-2007.5/5/23   | 2300*  | 10.1                         | 10.7   |
| 9-2001.5/5/23   | >2000*   | 9.4                          | >20.1  |
| 9-2003.5/5/23   | 2100   | 10.9                         | 21   |
| 7-2006/0/23   | 2100   | 14.5                         | 10   |
| 9-2006.5/5/23   | 2200   | 11.0                         | 21*  |
| 9-2003.5/5/23   | 2300   | 11.0                         | 21   |

\* Some loading rate of sample could not be taken due to specimen failure.

\* Experiment was terminated at 2000 psi, so the value of  
 "Ultimate Strain" is based on the ultimate stress.

pronounced decrease in ultimate stress and extended the rock salt ductility, i.e. the amount of rock salt deformation prior to a loss in load bearing ability (Table 1).

(7) Fracture at ambient temperature appears brittle and is followed by a near vertical post-failure curve up to approximately 250 psi (1.7 MPa) confining pressure. The slope of the post-failure curve in stress-strain space increased, i.e. becomes less negative, around and above 500 psi (3.5 MPa) confining pressure. At 200°C unconfined "failure" developed at  $(\sigma_1 - \sigma_3) = 2100$  psi (14.1 MPa) and  $\epsilon_1 = 12\%$ .

The foregoing summary of quasi-static experimental results is based primarily on approximately fifty triaxial experiments which were performed on 3.0 in (10 cm) diameter core specimens at Sandia Laboratories. It is supported by additional uniaxial experiments on 4.25 in (10.6 cm) diameter samples and on 2 in (5.1 cm) diameter samples by RF/SPEC, Inc.(9). Although most of these short-term data were remarkably reproducible both qualitatively and quantitatively, it must be recognized that only the unconfined data of RF/SPEC, Inc.(9) might be significant statistically. Shortage of like core made it impossible to repeat experiments in large numbers at all stresses and temperatures of interest. Furthermore the interpretation of experimental results remains incomplete unless mineralogical and textural variation of core are accounted for.

## CREEP EXPERIMENTS

### Apparatus and Experimental Procedures

All creep experiments were performed in new triaxial apparatus which accommodate relatively large cylindrical samples of up to 4.25 in (10.8 cm) diameter and 8.5 in (21.6 cm) in length. A schematic of this equipment is included as Fig. 1. Axial holes through the top of the pressure vessel and through the loading piston (Part 4) make it possible to vent the specimen by means of appropriate end-caps. Fig. 1 also indicates the use of multiple heaters to compensate for conduction heat losses through the floating base of the pressure vessel.

Testing is accomplished according to the following procedures. Each specimen is coated with a thin layer of MTV Silastic 108 (or equivalent), placed between two steel end-caps and enclosed in a Viton jacket. The specimen dimensions vary between 3.38 and 3.95 in (8.6 - 10.0 cm) in diameter by 8.5 to 11.3 in (21.6 - 28.7 cm) in length. In turn, the diameter of the end-caps is either 4.1 in or 4.3 in (10.4 - 10.9 cm) depending on the final sample diameter which is anticipated in any particular experiment.

To ensure good alignment all specimen ends are machined parallel to  $\pm 0.001$  in ( $\pm 0.03$  mm). End effects at the interface between rock salt and end-caps are minimized by polishing the end-cap faces to a mirror finish. Also, thin layers of molybdenum disulfide (Molykote) are provided between sample and end-caps.

Restrictions which are generated by the Viton jackets are small. For a typical wall thickness,  $t = 0.06$  in (1.5 mm), and for the standard internal jacket diameter,  $D = 4.0$  in (10.2 cm), 4 psi (34 kPa) of internal air pressure produce diameter changes of 12% at the midlength of the jacket and 27% at the rock/end-cap interface approximately 1 in (2.5 cm) away from a stiff clamp.

After jacketing, each specimen is placed inside the pressure vessel, pressurized hydrostatically and heated if desired, in that order. Heating is accomplished in 50°F increments over periods of two hours per temperature increment. After the target temperature is reached an additional soak period of no less than four hours is added. Overall, regardless of temperature, at least twelve hours elapse between the time of hydrostatic pressurization and the time at which a constant deviatoric stress is superimposed to initiate creep.

Deviatoric loading is achieved rapidly by opening a hand valve between the main loading ram (Part 2) in Fig. 1 and one of 5 gal (0.02 m<sup>3</sup>) precharged accumulators. The exact rate of deviatoric loading is determined in part by the magnitude of the fluid ram load (principal stress difference) and by the specimen compliance which depends on temperature. However, the loading rate is dictated primarily by the flow resistance of 1/4 in I.D. (6.4 mm) line tubing. Typically, the deviatoric loading rate falls into the range 3500 to 7000 psi/min (24.1 - 48.2 MPa/min).

Once a creep test is underway the ram load is maintained constant unless measurements of specimen area call for manual updates so as to maintain an approximately fixed value of principal stress difference. In turn, confining pressure is regulated intermittently by means of a servo-controlled pressure regulator (14).

In practice the number of system updates and therefore, the accuracy in the control of both  $(\sigma_1 - \sigma_3)$  and  $\sigma_3$  varies depending on the observed creep rates. It also depends sensitively on ambient temperature variations which influence the feed-back signal of the confining pressure system. In view of all these parameters, the range of optimum to worst control lies between

of 10, 15, 20, and 25 psi (0.7, 1.0, 1.4, and 1.7 MPa) the principal stress difference without confining pressure and 20 psi (1.4 MPa) in confining pressure are used to determine the specimen. Occasionally higher recoveries can be obtained depending on the experiment lasting less than 10 hours.

At the completion of any experiment where creep was sustained, the specimen is removed through a 1/2 inch opening and is removed at once. Following the removal, the specimen is in a state of hydrostatic swelling. At that point, in some cases attempts are made to record specimen recovery. Alternatively, hydrostatic swelling studies (PCCOE) are conducted following an experiment. The hydrostatic swelling may be as low as 10% for the procedure used.

Finally, at the end of each test, at elevated temperature, the specimen was removed and weighed and perhaps after 24 hours (in the case where we did not use confining pressure). In later experiments, sample swelling procedures are recommended. Following recovery in some cases which might be used to determine the final strain.

#### Instrumentation and Data Acquisition

The principal stress (principal stress difference) and confining pressure are maintained, and the pressure vessel. Friction effects are negligible. Various piezoelectric strain gauge instrumented transducers are employed, covered by water-cooled copper jackets. Standard force gauges with piezoelectric sensors are used as backups to measure gas pressure and confining pressure in the event of transducer malfunctions and/or electrical power failures.

Specimen temperature is given directly by the output of the control thermocouples of the two heaters which are shown in Fig. 1. This procedure proved accurate to within  $\pm 0.5^\circ\text{C}$  after calibrations established an accuracy

between the temperature at the thermocouple locations and the temperature over the entire length of the test piece.

To determine axial and radial strain (shear and volume deformation), the total axial sample shortening and the average radial "bulging" of each specimen are measured. The former is accomplished by means of an externally mounted linear variable differential transformer (LVDT). The latter measurement is made by tracking the volume (dilatometric) adjustments of the confining pressure fluid which are necessary to hold the confining pressure fixed (11). Details concerning all deformation measurements and a number of calibrations are described elsewhere (11). However, it is reiterated here that records of fluid volume changes are distorted by drift ( $\pm 0.5^\circ\text{C}$  in the ambient laboratory temperature). Although such drift usually is diurnal in nature and, therefore, identifiable in long-term creep tests, it can result in errors of as much as  $\pm 0.04\%$  radial strain. Errors of this magnitude can become overriding at low creep rates, less than approximately  $5 \times 10^{-9} \text{ s}^{-1}$ .

Data acquisition is accomplished by means of a multichannel, partially programmable electronic data logger (Interline Angus, model H2064) and a data terminal with cassette tape system (Texas Instruments Silent 700, 40K series with Vot tape unit). Five parameters are monitored at intervals of one hour or less: time, ram load (Fig. 1), confining pressure, LVDT signal and dilatometer output. All raw data are transferred at appropriate times onto a PDP 11/34 laboratory computer for reduction and display on a Tetrax 4010 graphics terminal.



#### MATERIAL TECHNIQUE

Detailed descriptions of New Mexico rock salt from drilled cores #1 and #177 were given in two references (16, 17). The core descriptions were obtained from geologist HHA #1, depth 1830 - 1100 ft (611 - 331 m) and #177, 2078 - 2091 ft (632 - 641 m). Typically the rock salt was made to coarse grades of 1 - 10 mm and 1 - 50 mm. Larger grains were encountered locally but have been avoided so far in this study. The grain size of all accessory minerals varied from very fine-grained (< 0.1 mm) to fine grained (0.1 - 1 mm). The average water content of rock salt core over the depths of interest was 1.6% (17).

Microscopic examinations (16) indicated that the upper level salt contained 64% normal per cent halite. Accessory mineral and their amounts were given as: polyhalite 1 - 6%, anhydrite tr-tr (tr indicates trace), and clay and silt nonconformable-silt. The lower level salt, HHA #1, 2078 - 2091 ft (632 - 641 m) and HHA #177, 2078 - 2091 ft (632 m) contained at least 64% normal per cent halite. Polyhalite amounted to tr-tr, anhydrite from tr-tr.

The foregoing microscopic description is in poor agreement with qualitative, macroscopic observations (16) which indicate high percentages of clay (up to 80%) in core from HHA #9, 2078 - 2091 ft (632 - 641 m) and in one deeper sample from 2620 ft (800 m). It is possible that the macroscopic records were influenced by discolorations due to hematite. On the other hand, the microscopic data reported are consistent with quantitative measurements (17) which identify HHA #5 rock salt, 2036 - 1070 ft (626 - 329 m) as more than 97 weight per cent water soluble with no more than 1.7% clay and silt. In samples from 2609 - 2618 ft (816 - 818 m) the water soluble content was > 96.1%. One specimen from 2705.6 ft (845 m) contained only 51.6% water solubles with 0.65% sulfates and 4.14% primarily clay and quartz (17). Water insoluble residues were negligible in several specimens between 2600 - 2611.5 ft (812 - 818 m).

#### THIS SUMMARY

The tests which are described in this report are listed in Table 1. The tests shown are similar to those completed by HF/HAC, Inc. for rock salt from the lower horizon of the WMS study area (10, 11). However, instead of duplicating experimental conditions throughout for like core, most tests actually complement each other in virtue of overlapping but not necessarily identical parameter conditions. For example, earlier HF/HAC experiments were carried out at principal stress differences of 1500, 3000 and 4,500 psi (10.3, 20.6 and 31.6 MPa) (10, 11). The present tests on salt from below 2000 ft (610 m) were performed at approximately 1000, 2200 psi (6.9, 15.2 MPa), etc. Beyond this examination of efforts there exist some important distinctions. The tests of this study provide data on rock salt from 317 to both 1000 = 2100 ft (305 = 605 m). Also, core from 2000 = 2200 ft (610 = 671 m) was tested at the low stress difference of 1000 psi (6.9 MPa) and up to a peak temperature of 300°C. Finally, larger samples were employed and the test durations were increased to an average of 600 hours per experiment. Both facts enhance the resolution of creep rates particularly around or below  $10^{-9}$  s<sup>-1</sup>.

TABLE 2  
Creep Test Matrix

| Specimen<br>No.                                       | $\sigma_1 - \sigma_3$<br>(psi) (MPa) |      | $\sigma_3$<br>(psi) (MPa) |      | $\dot{\epsilon}$<br>(%) |
|---|--------------------------------------|------|---------------------------|------|-------------------------|
| Upper Level Rock Salt, Depth 2000-2100 ft (625-650 m) |                                      |      |                           |      |                         |
| 1   | 2000                                 | 13.8 | 500                       | 3.5  | 22                      |
|   | 4500                                 | 31.0 | 500                       | 3.5  | 22                      |
| 2   | 3000                                 | 20.7 | 500                       | 3.5  | 22                      |
|   | 4500                                 | 31.0 | 500                       | 3.5  | 22                      |
| 3   | 2000                                 | 13.8 | 3000                      | 20.7 | 22                      |
|   | 4500                                 | 31.0 | 3000                      | 20.7 | 22                      |
| 4   | 3000                                 | 20.7 | 3000                      | 20.7 | 22                      |
|   | 4500                                 | 31.0 | 3000                      | 20.7 | 22                      |
|   | 3000                                 | 20.7 | 3000                      | 20.7 | 22                      |
| Lower Level Rock Salt, Depth 2600-2800 ft (810-875 m) |                                      |      |                           |      |                         |
| 5   | 1000                                 | 6.9  | 500                       | 3.5  | 22                      |
|   | 2200                                 | 15.2 | 500                       | 3.5  | 22                      |
|   | 3300                                 | 22.8 | 500                       | 3.5  | 22                      |
|   | 2200                                 | 15.2 | 500                       | 3.5  | 22                      |
| 6   | 1000                                 | 6.9  | 3000                      | 20.7 | 22                      |
| 7   | 1000                                 | 6.9  | 0                         | 0    | 100                     |
| 8   | 1000                                 | 6.9  | 500                       | 3.5  | 100                     |
| 9   | 1000                                 | 6.9  | 500                       | 3.5  | 100                     |
|   | 2200                                 | 15.2 | 500                       | 3.5  | 100                     |
| 10  | 1000                                 | 6.9  | 3000                      | 20.7 | 100                     |
|   | 2200                                 | 15.2 | 3000                      | 20.7 | 100                     |
| 11  | 1000                                 | 6.9  | 500                       | 3.5  | 200                     |
| 12  | 1000                                 | 6.9  | 3000                      | 20.7 | 200                     |
| 13  | 4500                                 | 31.0 | 500                       | 3.5  | 22                      |

### EXPERIAL

Table 1 contains a summary of general statistics of all creep experiments, including stress states, test durations and numerically greatest axial and radial strains. The actual stresses which were obtained here differ slightly from the target stresses in Table 2. The highest values of principal stress difference represent the following fractions of the ultimate stress values  $\sigma_u$  which were measured at a loading rate  $d(\sigma_1 - \sigma_3)/dt = 30 - 60$  psi/min (0.21 - 0.41 MPa/min) (Table 1):

$$\begin{aligned}(\sigma_1 - \sigma_3) &= 0.55\sigma_u \text{ at } \sigma_3 = 500 \text{ psi (3.5 MPa), } T = 22^\circ\text{C} \\(\sigma_1 - \sigma_3) &= 0.75\sigma_u \text{ at } \sigma_3 = 500 \text{ psi (3.5 MPa), } T = 105^\circ\text{C} \\(\sigma_1 - \sigma_3) &= 0.43\sigma_u \text{ at } \sigma_3 = 500 \text{ psi (3.5 MPa), } T = 200^\circ\text{C} \\(\sigma_1 - \sigma_3) &\leq 0.26\sigma_u \text{ at } \sigma_3 = 3000 \text{ psi (20.7 MPa), } T = 22^\circ\text{C} \\(\sigma_1 - \sigma_3) &\leq 0.26\sigma_u \text{ at } \sigma_3 = 3000 \text{ psi (20.7 MPa), } T = 200^\circ\text{C}\end{aligned}$$

Fig. 1 shows typical records of  $(\sigma_1 - \sigma_3)$  versus time. Relatively small cyclical variations in stresses are due to diurnal changes in the ambient laboratory temperature ( $\pm 0.5^\circ\text{C}$ ).

Test durations varied from as little as 40 hours in one experiment at  $200^\circ\text{C}$  to 1542 hours in an ambient temperature test. The average test duration was 665 hours.

Table 3 also lists the total axial and radial creep strains which were reached in each experiment. Throughout, strains are given as engineering strains, e.g.  $\Delta L/L$ , relative to the sample dimensions at the beginning of each test. Note that some radial strain data are marked as uncertain either because data analyses have not been completed or because they require additional checks. For the same reason the observed volumetric strains are described only in a semi-quantitative sense. Small amounts of compaction which are indicated in Table 3 may or may not prove to be real. In any event, the total volumetric creep corresponding to the first stages of tests 9-2063.5 are small ( $\epsilon > -1\%$ ). Generally, the same appears to apply to all tests where Table 3 now shows question marks.

TABLE 3

## Summary Statistics of Creep Tests

Stress and strain represent true stress and engineering strain, respectively.

 $\epsilon_1$  : axial strain;  $\epsilon_2$  : radial strain;  $\epsilon_3$  : volumetric strain. Parentheses denote uncertainties.

Question marks indicate that data are missing or not yet available.

Key - DDD : large dilatancy (&gt; 4%); DD : considerable dilatancy (&gt; 1%); D : some dilatancy (&lt; 1%); C : some compaction (&lt; 0).

| Sample<br>(Hole# - Depth (ft)) | $\sigma_1 - \sigma_3$<br>(psi) | $\sigma_3$<br>(MPa) | $\epsilon_3$<br>(psi) | $\epsilon_3$<br>(MPa) | Temper.<br>(°C) | Test Duration<br>(Hrs) | $(\epsilon_1)$ max<br>(%) | $(-\epsilon_2)$ max<br>(%) | Direction<br>of $\epsilon$ |
|--------------------------------|--------------------------------|---------------------|-----------------------|-----------------------|-----------------|------------------------|---------------------------|----------------------------|----------------------------|
| Upper Level Salt               |                                |                     |                       |                       |                 |                        |                           |                            |                            |
| 9-2078                         | 2000                           | 13.8                | 500                   | 3.5                   | 22              | 310                    | .273                      | .169                       | C                          |
| 9-2078                         | 4500                           | 31.0                | 500                   | 3.5                   | 22              | 166                    | 7.18                      | 4.57                       | DD                         |
| 9-2083                         | 2950                           | 20.3                | 500                   | 3.5                   | 22              | 1238                   | 2.22                      | 1.68                       | DD                         |
| 9-2083                         | 4500                           | 31.0                | 500                   | 3.5                   | 22              | 360                    | 13.80                     | >0.58                      | DDD                        |
| 9-2078.5                       | 1700                           | 11.7                | 3000                  | 20.7                  | 22              | 311                    | .125                      | (0.02)                     | ?                          |
| 9-2078.5                       | 4100                           | 28.3                | 3000                  | 20.7                  | 22              | 166                    | 3.62                      | 2.78                       | C                          |
| 9-2083.5                       | 2900                           | 20.0                | 3000                  | 20.7                  | 22              | 1035                   | 2.12                      | ?                          | (C)                        |
| 9-2083.5                       | 4400                           | 30.3                | 3000                  | 20.7                  | 22              | 1256                   | 11.40                     | (>6.4)                     | (D)                        |
| 9-2083.5                       | 3000                           | 20.7                | 3000                  | 20.7                  | 22              | 675                    | 3.34                      | ?                          | ?                          |
| Lower Level Salt               |                                |                     |                       |                       |                 |                        |                           |                            |                            |
| 9-2625                         | 1200                           | 8.3                 | 500                   | 3.5                   | 22              | 1842                   | .22                       | .06                        | C                          |
| 9-2625                         | 2200                           | 15.2                | 500                   | 3.5                   | 22              | 1481                   | 2.27                      | (1.30)                     | D                          |
| 9-2625                         | 2300                           | 22.8                | 500                   | 3.5                   | 22              | 356                    | 5.31                      | (2.25)                     | ?                          |
| 9-2625                         | 2300                           | 15.9                | 500                   | 3.5                   | 22              | 595                    | 0.03                      | (0.07)                     | C                          |
| 9-2677                         | 1150                           | 7.9                 | 3000                  | 20.7                  | 22              | 671                    | 0.15                      | (0.06)                     | (C)                        |
| 9-2672.5                       | 1000                           | 6.9                 | 30                    | 0.2                   | 100             | 612                    | 2.94                      | ?                          | ?                          |
| 9-2624                         | 1050                           | 7.3                 | 500                   | 3.5                   | 100             | 652                    | 2.73                      | 1.50                       | ?                          |
| 9-2686                         | 1000                           | 6.9                 | 500                   | 3.5                   | 100             | 747                    | 0.83                      | 0.17                       | (C)                        |
| 9-2686                         | 2350                           | 16.2                | 500                   | 3.5                   | 100             | 190                    | 17.6                      | 11.40                      | ?                          |
| 9-2671                         | 930                            | 6.4                 | 3030                  | 20.9                  | 100             | 1174                   | 0.98                      | (0.48)                     | ?                          |
| 9-2671                         | 2250                           | 15.9                | 3000                  | 20.7                  | 100             | 568                    | 29.3                      | (10.6)                     | ?                          |
| 9-2688                         | 1000                           | 6.9                 | 530                   | 3.7                   | 200             | 80                     | 6.43                      | 1.74                       | ?                          |
| 9-2668                         | 1000                           | 6.9                 | 3000                  | 20.7                  | 200             | 160                    | 10.18                     | 1.5                        | ?                          |
| 9-2777                         | 4800                           | 33.1                | 500                   | 3.5                   | 22              | 100                    | 17.7                      | 14.5                       | DDD                        |

The large values of dilatancy,  $\epsilon < -4\%$ , of experiments 9-2083, 9-2686, and 9-2777, are associated with accelerated creep and impending creep fracture. It should be noticed that the total axial and radial creep strains in the latter tests are bounded by fracture strain values which were measured in short-term, quasi-static tests into the post-failure regime.

Files of representative data for all experiments are compiled in an appendix. For the time being these files are restricted to axial strain time data  $(\epsilon_1, t)$ . Obviously, the number of data points (5 to 12) in these condensed files is much smaller than in the original files (typically 750). Indications of data scatter are included deliberately to demonstrate maximum variations in the creep records. These variations are caused simply by experimental scatter or by the number of updates of ram force (Fig. 2) to maintain a mean constant principal stress difference. The importance of these updates is particularly evident in Figs. 3 and 4 for specimen 9-2777. Here updates in ram force were delayed deliberately to establish likely differences in results between constant force tests and constant stress (creep) experiments. Clearly, these differences can be significant. Constant force tests may alter the course of primary creep at constant stress and may prevent the development of secondary creep altogether.

Figs. 5 through 12 contain examples of crossplots to explore the effects of principal stress difference, temperature and confining pressure. Indications of the reproducibility of results are provided by three sets of data, in Figs. 5 through 8 and in the digitized data files 9-2078/.5/22 and 9-2083/.5/22, 9-2078.5/3/.22 and 9-2083.5/3/.22 and 9-2624/.5/100 with 9-2686/.5/100 (Appendix). The large differences in results of experiments 9-2624/.5/100 and 9-2686/.5/100 are unexplained but deemed exceptional.

The comparisons in Figs. 5 through 12 assume that strain history has no major influence on creep as stress difference is increased. Therefore, no distinction was made between experiments which were performed on the same

sample if the specimen was subjected to successively increasing levels of principal stress difference. Strain history effects with decreasing principal stress difference will be discussed later.

A cursory comparison of the results for rock salt from both formations of interest to the WIPP (2000 - 2100 ft and 2650 - 2800 ft) indicates that trends in creep behavior are independent of depth. Therefore, the following discussion will not consider depth as a variable although the data pertaining to the upper and lower levels (Table 2) will not actually be mixed. The general agreement in results is consistent with the high purity of salt tested (i.e., 99.99%).

#### General Features

The creep plots, Figs. 5 through 12, demonstrate that New Mexico rock salt undergoes the three classical stages of creep: primary (transient) creep, secondary (steady state) creep and tertiary (accelerated) creep which terminates in creep fracture. Limited radial strain data (Figs. 6 and 8) further indicate that creep appears to proceed with little change in rock volume at low principal stress difference, small strains and/or high temperature. In turn, noticeable dilatancy develops as  $(\sigma_1 - \sigma_3)$  is raised. Eventually large amounts of dilatancy (damage) led into creep fracture.

In a qualitative sense the creep data shown verify the well known non-linear effects of principal stress difference and temperature (Fig. 4 through 12). On the other hand, first examinations in this study suggest that pressure is important only in the control of tertiary creep and creep fracture.

Periodic measurements on deformed samples after testing led to the conclusion that essentially all creep strain was irreversible. This conclusion was strengthened by monitoring the axial and radial dimensions of two specimens continuously for periods of twenty days. No recovery was resolved either by

means of dial gauges (axial strain) or by means of strain gauge instruments clip-on gauges (radial strain).

### Primary (Transient) Creep

Efforts were initiated to describe primary creep of New Mexico rock salt objectively. Preliminary work consisted of fitting axial strain-time data ( $\epsilon_p$ ,  $t$ ) to logarithmic and power relations of the form

$$\epsilon_p' = A \log(t) \quad (i)$$

$$\epsilon_p'' = Ct^n \quad (ii)$$

where A, C, and n are constants. Typical log-log and semi-log plots of the data are given in Figs. 15 and 16. Although the foregoing equations are empirical they have been used widely and often provide excellent approximations to experimental data at constant stress. These equations are not constitutive equations, and it is recognized that equations (i) and (ii) predict infinite strain rates at zero time. Obviously the interpretation of Figs. 15 and 16 requires considerable judgment. For this preliminary analysis two assumptions were made which permit comparisons of the present results among one another and with published primary creep data for rock salt. First, primary creep,  $\epsilon_p$  and secondary creep  $\epsilon_{ss}$ , proceed simultaneously such that  $\epsilon_{total} = \epsilon_p + \epsilon_{ss}$  (14). It was also assumed that total creep is governed by primary creep at less than approximately 60 hours.

In some cases (Fig. 16) it appeared that primary creep might be described best as logarithmic creep. However, in the majority of experiments maximum statistical correlations were obtained by means of power relations which are summarized in Table 4 in terms of the "constants" C and n. For the sake of



Summary of Primary (Transient) Creep Rate According  
to Data Fit of Type  $\dot{\epsilon}_1 = C\sigma^n(t/t_0)^{-m}$ ;  $t$  : time (s);  $l = 1, 2$   
Stress and strain represent true stress and engineering strain, respectively.

19

consistency all fits were made within the time interval  $1 \leq t \text{ (hrs)} \leq 50$ . With three exceptions the selection of this time interval ensured the highest correlations between actual data and the approximations by means of equation (10). Combinations of logarithmic and power creep, e.g.  $\epsilon_p = \epsilon_p' + \epsilon_p''$ , are possible but have not yet been considered in detail.

Although further analyses are needed, two observations deserve mentioning. (1) If a "power law" creep formulation is accurate for constant stress, then the multiplier  $C$  and the time exponent  $n$  appear to be variable. According to Table 4,  $n$  increases up to a factor of five with increases in principal stress difference and temperature. Considerable differences in the values of  $n$  remain even if allowances are made for the influence of secondary creep particularly at elevated temperature. More work is needed to determine whether the parameters  $C$  and  $n$  are independent of one another. Coupling between  $C$  and  $n$  was indicated by some earlier data for artificial and natural rock salt (3, 10, 11). (2) The influence of confining pressure on axial creep ( $\epsilon_1, t$ ) appears to be small, particularly at low  $(\sigma_1 - \sigma_3)$ . This follows from comparisons of the empirical variables  $C$  and  $n$ . It also follows from a comparison of normalized axial strain data,  $\epsilon_1/\epsilon_1$ , which were calculated from  $\epsilon_1 = C t^n$  at  $t = 100$  hours (Table 4). However, caution should be exercised in generalizing this result to shear strain  $(\epsilon_1 - \epsilon_3)$ , i.e. before considering further the nature of radial creep.

#### Secondary (Steady State) Creep

The existence of secondary creep is difficult to establish without doubt. In this case, estimates of secondary creep rates in Table 5 are based on two criteria. First, plots had to contain what "looked like" straight line portions. More objectively, it was required that the slope  $r$  of plots in the space  $\log(\epsilon_{\text{total}}), \log(t)$  increased with time from the value  $r = n$  during primary

TABLE 5

Summary of Secondary Creep Estimates  
 (Parentheses denote upper bound values)  
 Stress and strain represent true stress and engineering strain, respectively.

| Sample<br>(Hole# - Depth (ft)) | $\sigma_1 - \sigma_2$<br>(psi) (MPa) | $\sigma_3$<br>(psi) (MPa) | Temper.<br>(°C) | Estimated<br>Secondary Creep<br>Rate $\dot{\epsilon}_1$ ( $10^{-8}$ /s) |     |         |
|--------------------------------|--------------------------------------|---------------------------|-----------------|---|-----|---------|
| Upper Level Salt               |                                      |                           |                 |   |     |         |
| 9-2076                         | 2000                                 | 13.8                      | 500             | 3.5   | 22  | (0.042) |
| 9-2076                         | 4500                                 | 31.0                      | 500             | 3.5   | 22  | (2.02)  |
| 9-2083                         | 2950                                 | 20.3                      | 500             | 3.5   | 22  | (0.16)  |
| 9-2083                         | 4500                                 | 31.0                      | 500             | 3.5   | 22  | (5.8)   |
| 9-2078.5                       | 1700                                 | 11.7                      | 3000            | 20.7  | 22  | (0.012) |
| 9-2078.5                       | 4100                                 | 28.3                      | 3000            | 20.7  | 22  | (2.25)  |
| 9-2083.5                       | 2900                                 | 20.0                      | 3000            | 20.7  | 22  | (0.23)  |
| 9-2083.5                       | 4400                                 | 30.3                      | 3000            | 20.7  | 22  | (1.10)  |
| 9-2083.5                       | 3000                                 | 20.7                      | 3000            | 20.7  | 22  | 0.018   |
| Lower Level Salt               |                                      |                           |                 |   |     |         |
| 9-2625                         | 1000                                 | 6.9                       | 500             | 3.5   | 22  | (0.014) |
| 9-2625                         | 2500                                 | 15.2                      | 500             | 3.5   | 22  | (0.20)  |
| 9-2625                         | 4400                                 | 29.8                      | 500             | 3.5   | 22  | (1.93)  |
| 9-2625                         | 2200                                 | 15.0                      | 500             | 3.5   | 22  | (0.033) |
| 9-2671                         | 1150                                 | 7.9                       | 3000            | 20.7  | 22  | (0.008) |
| 9-2672.5                       | 1000                                 | 6.9                       | 30              | 0.2   | 100 | 1.00    |
| 9-2628                         | 1000                                 | 7.2                       | 500             | 3.5   | 100 | 0.61    |
| 9-2666                         | 1000                                 | 6.9                       | 500             | 3.5   | 100 | 0.25    |
| 9-2666                         | 2000                                 | 13.8                      | 500             | 3.5   | 100 | 0.96    |
| 9-2671                         | 1000                                 | 6.9                       | 3000            | 20.7  | 100 | 0.18    |
| 9-2671                         | 2000                                 | 13.8                      | 3000            | 20.7  | 100 | 3.41    |
| 9-2668                         | 1000                                 | 6.9                       | 500             | 3.5   | 200 | 14.0    |
| 9-2668                         | 1000                                 | 6.9                       | 3000            | 20.7  | 200 | (14.7)  |
| 9-2717                         | 4000                                 | 27.1                      | 500             | 3.5   | 22  | 31.20   |

creep (Table 4) towards  $n = 1$  when secondary creep becomes overriding. For  $\epsilon_{\text{total}} = \epsilon_1 + \epsilon_{\text{ss}}$ , rising values of  $n$  indicate that the contribution of secondary creep to the total creep strain is increasing. If the second criterion was not satisfied clearly, estimates of creep rates in Table 5 are the smallest observed creep rates at the end of each test. Hence, they only concern the upper bounds and are listed in parentheses.

To compare the secondary creep results of New Mexico rock salt quickly, the assumptions were made that the total creep is thermally activated, that the effect of stress, temperature and pressure are separable and that the temperature dependence can be described by a single activation energy. It was further stipulated that secondary creep rates are proportional to a power function of stress (16-20). Then, following standard procedures the graphs of Figs. 17 - 19 were constructed for upper and lower level salt. Figs. 17 and 18 include data for 2 in. diameter samples of BR/SPCC, Inc. (10, 11). Collecting the approximations in some of the results in parentheses, estimated values of activation energy fall into the relatively narrow range between  $Q = 5.8$  kcal/mole and  $Q = 13.6$  kcal/mole. The number of data available does not suffice to determine possible variations in  $Q$  with temperature. A pressure effect is weak if it is present at all.

Stress exponents of the results in Table 5 are listed in Figs. 17 and 18 for both salt horizons considered. Keeping different groups of tests deliberately apart, the stress exponents vary considerably between 3.1 and 6.5 depending on temperature and which results are included in a particular linear data fit. Well defined secondary creep rates correspond to  $n = 3.1$  at  $100^\circ\text{C}$ . The remaining values only correspond to secondary creep estimates (smallest observed creep rates).

#### Tertiary Creep and Creep Fracture

Tertiary creep was measured in four experiments at 500 psi (3.5 MPa) confining pressure and  $T = 22^\circ\text{C}$  and at  $T = 100^\circ\text{C}$  with confining pressures of

0, 500 and 3000 psi (0, 3.5, 20.7 MPa). Except in one case,  $\sigma_3 = 3000$  psi (20.7 MPa; sample 9-2671), tertiary creep terminated in creep fracture associated with a considerable amount of dilatancy much like that which is measured at the ultimate stress in quasi-static tests. At 3000 psi (20.7 MPa) accelerated creep started at  $\epsilon_1 \approx 13.5\%$  and continued to  $\epsilon_1 \approx 29\%$  over several hundred hours and with an approximately 10-fold increase in creep rate. However, no fracture was imminent at the peak strain before the experiment was terminated for inspection of the specimen.

#### Effect of Stress or Strain History

It was stated in an earlier section that the data representations of Tables 4 and 5 implied no effect of stress or strain history under increasing stress difference. This assumption appears acceptable in the light of published creep results (5, 10, 11) but questionable in view of recent short-term data for variable stress paths which may or may not be relevant for creep (12). To evaluate the influence of more complicated histories and partly to check the accuracy of some secondary creep rates in Table 5, two sets of tests were performed in successive stages of increasing and decreasing stress. For one experiment (9-2683.5; Fig. 13) the principal stress difference was changed from 2500 psi (20.2 MPa for 1088 hours) to 4400 psi (30.3 MPa for 1256 hours) and back to 3000 psi (20.7 MPa for 675 hours). In the second experiment (9-2723; Fig. 14) ( $\sigma_1 - \sigma_3$ ) was varied from 2200 psi (15.2 MPa for 1451 hours) to 3300 psi (22.7 MPa for 356 hours) and again to 2300 psi (15.9 MPa for 595 hours). Details of these tests are listed in Tables 3-5. The most important result is that the estimated secondary creep rates at approximately 3000 psi (20.7 MPa) and 2300 psi (15.9 MPa) dropped by nearly one order of magnitude after creep had occurred at higher values of principal stress difference, i.e., 4400 (30.3 MPa) and 3300 psi (22.7 MPa), respectively (stages C2 and C3 in the captions of Figs. 13 and 14).

#### PRELIMINARY COMPARISON OF RESULTS WITH RE/SPEC DATA AND DISCUSSION

In view of limited data and data scatter it is important to compare the present results with the results of RE/SPEC Inc. (10, 11). Given the complementary nature of all tests, satisfactory agreement would automatically broaden the overall data base available for constitutive modeling and model validation.

Qualitatively, available results demonstrate that New Mexico rock salt undergoes all stages of creep from primary creep through creep fracture. More important, all of these stages were observed at levels of stress and temperature which are of interest to the WIPP. This means that past modeling efforts which have concentrated on the prediction of primary creep must be extended.

Primary and secondary creep are recognized readily in a qualitative sense. However, the combined data of Sandia and RE/SPEC Inc. indicate independently that unambiguous descriptions of these phenomena are difficult. Particular difficulties arise because primary and secondary creep appear to be overlapping, because primary creep at small time probably depends upon the initial loading rate and because most secondary creep rates are only upper bounds. As a result, the description of primary creep in terms of power creep, for example (Table 4), is sensitive to the time interval which is chosen to fit the experimental data in the space  $\log(\epsilon_1), \log(t)$ . Furthermore, the power creep relation which was used here renders poor fits to experimental measurements at times less than approximately one hour. At elevated temperature primary creep interpretations are distorted by fast secondary creep which is going on simultaneously. Whatever the reason, ambiguity in the interpretation of creep data probably produced errors in description and predictions. For example, based on Tables 4 and 5 the predicted total axial creep strain for sample 2-2671,  $(\sigma_1 - \sigma_2) = 2250$  psi (15.5 MPa) is 16.3% after 500 hours if  $\epsilon_{\text{total}} = \epsilon_p + \epsilon_{ss}$  as suggested by the nature of double logarithmic plots. This value overpredicts the actual strain accumulation ( $\epsilon_1 = 12.5\%$ ) after that time by approximately 30%.

Some qualitative disagreement between various sets of results arises concerning the influence of pressure. Available data (11, 12) led to the common conclusion that pressure enters as a first order effect into the prediction of tertiary creep and creep fracture. However, preliminary analyses of the present results suggest that the influence of pressure on transient and steady state creep is negligible. This observation runs contrary to earlier RE/CFEC data and to the results of short-term (quasi-static) experiments. To clarify this point more detailed examinations should be made of existing radial strain data. Also, further tests should be conducted particularly below 500 psi (3.5 MPa) confining pressure.

Preliminary quantitative comparisons of Sandia and RE/CFEC data are not totally satisfactory. One power creep description of RE/CFEC Inc. (11) appears to predict considerably larger creep strains than those which are indicated by the results of Table 4 and 5. For example, for primary creep periods of 100 hours several predictions at common stresses and temperatures differ by factors of approximately ten. RE/CFEC results also suggest higher secondary creep rates. However, it is emphasized that most of these comparisons have been carried out quickly and require checks and confirmation. Some of the observed differences might be due to specimen size. Values of average activation energies are relatively consistent and fall into the range  $7.2 \leq E_a \leq 13.6$  kcal/mole (Fig. 17). The majority of data fit  $E_a = 13.3$  kcal/mole.

It is not clear at this point how serious the observations of history effects might be if they are proven typical under decreasing principal stress difference. However, history effects might influence stress calculations and therefore affect the likelihood of creep fracture.

Tertiary creep and triaxial creep fracture developed at 170°C and surprisingly low values of principal stress difference ( $\sigma_1 - \sigma_3 = 220$  psi (15.0 MPa) even at  $\sigma_3 \geq 500$  psi (3.5 MPa) confining pressure. Although these stresses and temperatures are likely to develop only very locally (21), the

phenomena are significant enough to require additional study. For lack of established theories it will be necessary in part to conduct very long experiments. However, creep fracture predictions may also be aided by empirical correlations between creep fracture strains and quasi-static complete stress-strain characteristics (22, 23). Triaxial tests on granite, sandstone and marble suggested that complete quasi-static stress-strain records can be used to establish loci of limiting stable creep strains. Specifically, it appears that the total nonelastic (time-dependent) strain at any stress will not exceed the nonelastic strain to fracture at the same confining pressure and principal stress difference in quasi-static tests. Although this procedure is strictly empirical it appeals to the notion that nonelastic strains are a measure of damage and that the limiting strain establishes a maximum amount of damage as a function of stress state before an instability occurs with associated loss in load bearing ability. The foregoing empirical predictive scheme for creep fracture is invoked here because it correlates remarkably well with the available creep fracture observations in this report.



## SUMMARY

An interim summary of triaxial creep experiments on rock salt was presented which were carried out at Sandia Laboratories in support of the proposed Waste Isolation Pilot Plant (WIPP) in southeastern New Mexico. Tests were performed on relatively large samples measuring 3.9 in (9.9 cm) diameter. Details of relevant experimental procedures and measurements were described. Results shown include digital data in condensed data files and numerous graphical comparisons for rock salt from two horizons 2000 - 2100 ft and 1600 - 1700 ft (610 - 670 m; 410 - 575 m) as a function of principal stress difference, temperature and confining pressure. Qualitative examinations showed that the rock salt exhibited all known stages of creep, i.e. primary, secondary and tertiary creep followed by creep fracture.

Preliminary evaluations of axial strain time records during primary creep ( $\epsilon_1$ ,  $t$ ) are discussed assuming the applicability of so-called power "law" creep but without implying that the conventional power relationship is a constitutive equation. Contrary to earlier results, the primary creep data of this study appear to be independent of pressure within fairly wide experimental scatter. Estimates of steady state creep rates of interest to the WIPP vary from  $\dot{\epsilon}_1 < 10^{-10} \text{ s}^{-1}$  to approximately  $10^{-8} \text{ s}^{-1}$ . Tertiary creep and creep fracture were observed at ambient temperature and at 100°C. Creep fracture was not observed in a limited number of experiments at 200°C and is deemed unlikely except, possibly, at very low confining pressure,  $\sigma_3 \leq 100 \text{ psi}$  (0.7 MPa). To predict creep fracture the use of an empirical procedure was suggested which correlates permanent time-dependent strains (damage) with limiting damage which is established from complete quasi-static stress-strain curves including pre- and post-failure records.

The analysis of tests which are described is continuing and includes examinations of existing radial strain data. It also includes a definitive comparison of data with earlier results of Ht/SPEC Inc. beyond a cursory comparison which is included in this report.

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## PICTURES

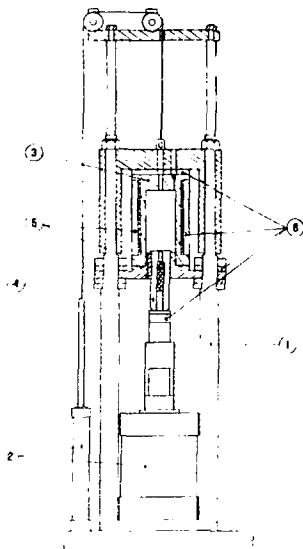


Fig. 1

Triaxial compression apparatus

Parts key: (1) load frame (2) hydraulic actuators  
(3) pressure vessel (4) loading ram (5) heaters  
(6) insulation

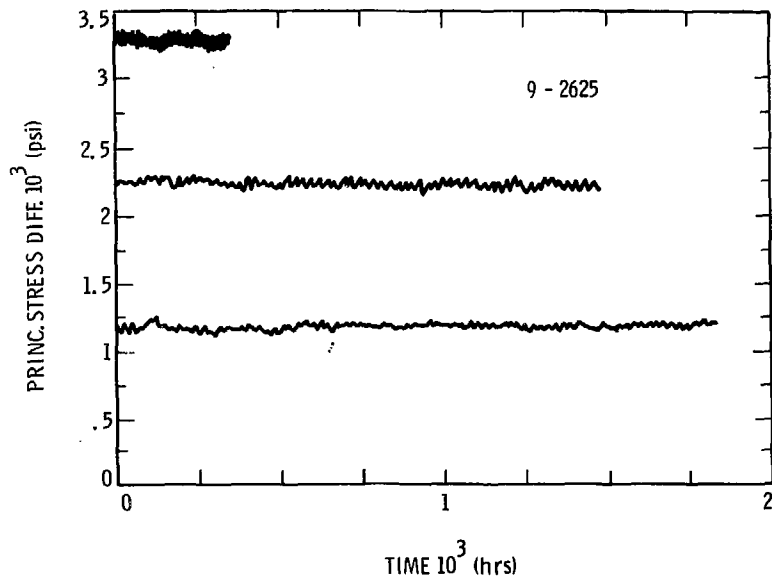


Fig. 2

Typical variations of principal stress difference with time for creep tests at 500 psi (3.5 MPa) confining pressure (Sample 9-2625).



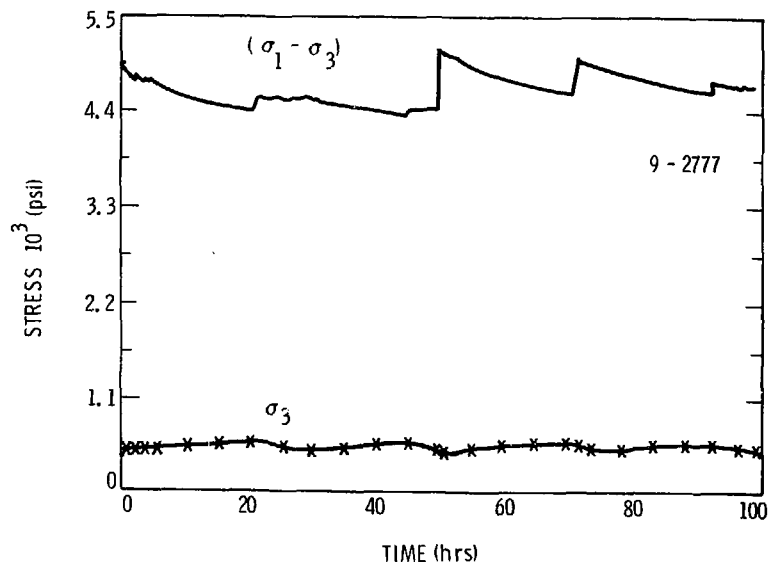


Fig. 3

Plot of principal stress difference and confining pressure versus time for sample 9-2777. Note extreme variations in  $(\sigma_1 - \sigma_3)$  during periods of constant applied ram load (Fig. 1).

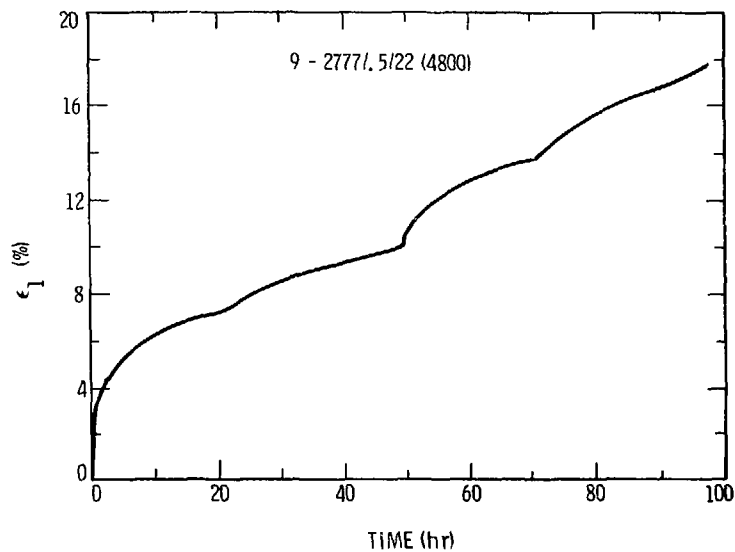


Fig. 4

Axial creep record ( $\epsilon_L$  vs.  $t$ ) for sample 9-2777L. Note strong influence of upsets in ( $\sigma_1 - \sigma_3$ ).

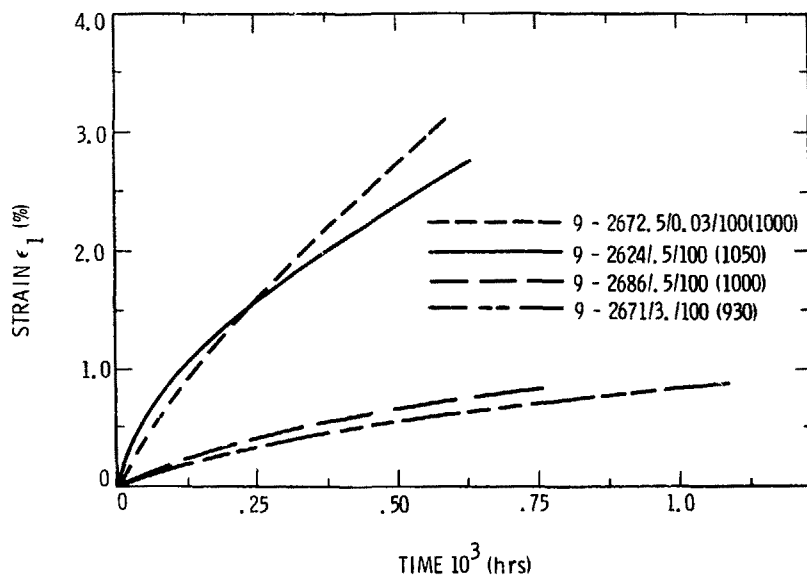


Fig. 1

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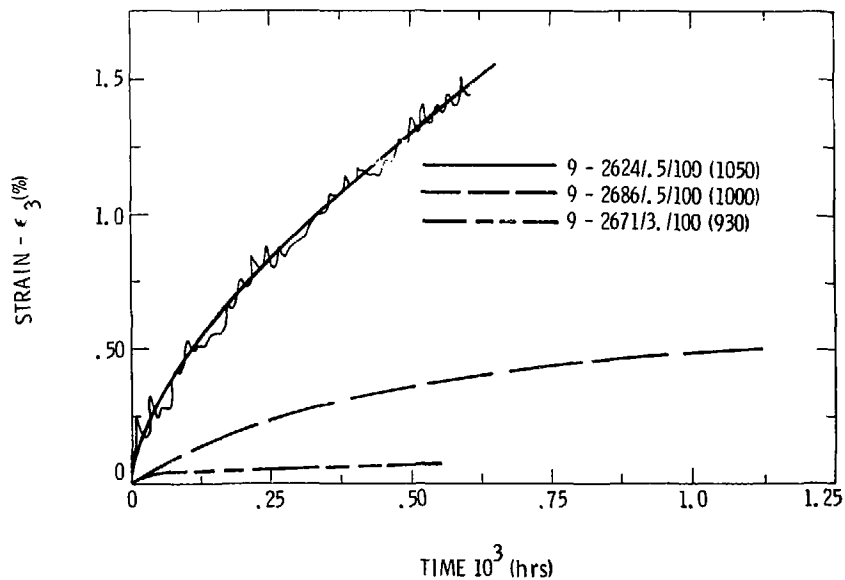


Fig. 4

Radial creep records ( $\epsilon_3$  vs.  $t$ ) at  $(\sigma_1 - \sigma_3) = 1000$  psi (6.0 MPa),  $T = 100^\circ\text{C}$  and variable confining pressure (for test code see Fig. 5).

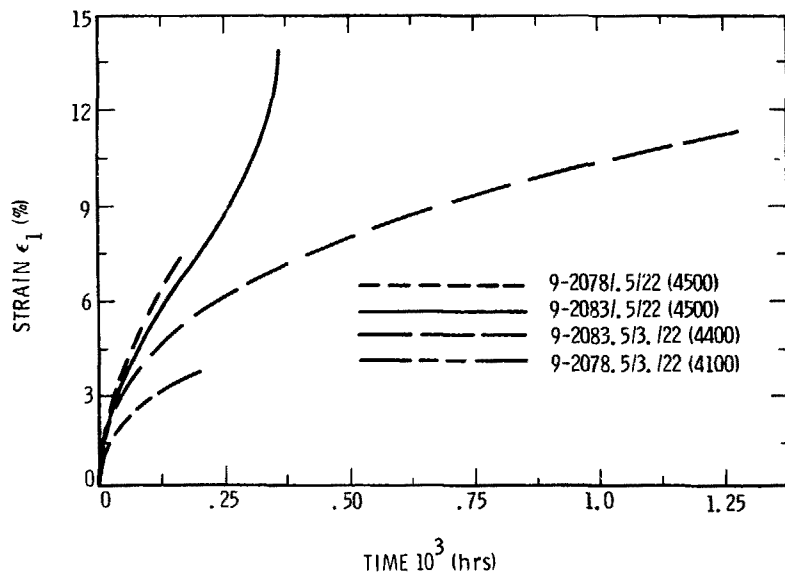


Fig. 7  
Axial creep records ( $\epsilon_1$  vs.  $t$ ) at  $(\sigma_1 - \sigma_3) = 4500$  psi (31.0 Mpa),  $T = 70^\circ\text{C}$  and variable confining pressure (for test code see Fig. 5).

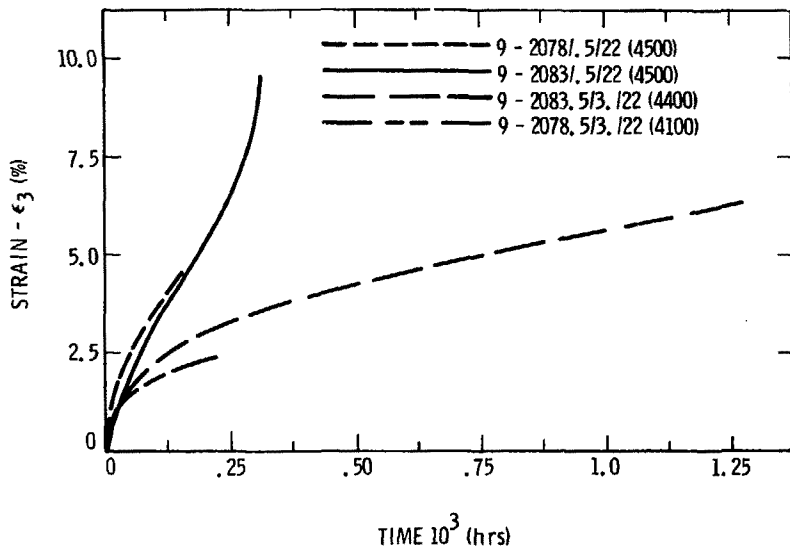


Fig. 8

Radial creep records ( $-\epsilon_3$  vs.  $t$ ) at  $(\sigma_1 - \sigma_3) = 4500$  psi (31.0 MPa),  $T = 22^\circ\text{C}$  and variable confining pressure (for test code see Fig. 5).

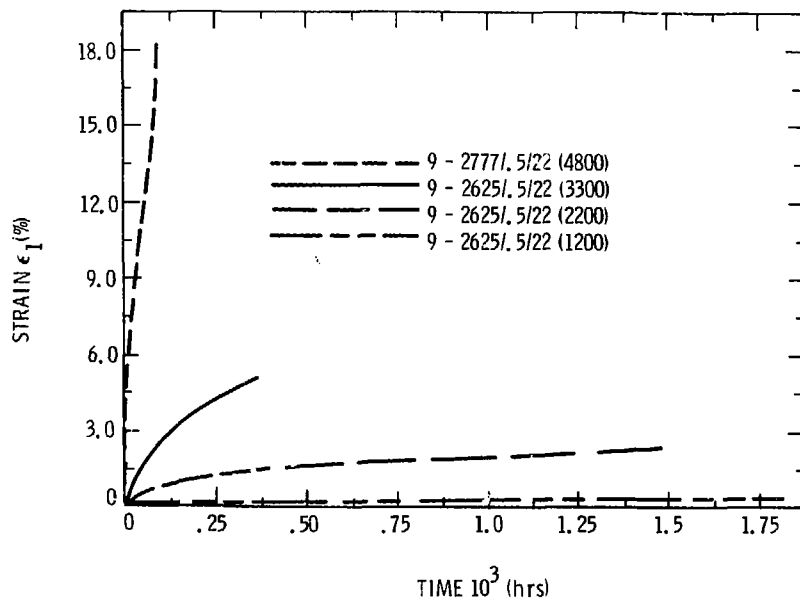


Fig. 9

Axial creep records ( $\epsilon_1$  vs.  $t$ ) at  $\sigma_3 = 500$  psi (3.5 MPa),  
 $T = 22^\circ\text{C}$  and variable principal stress difference (for  
 test code see Fig. 5).

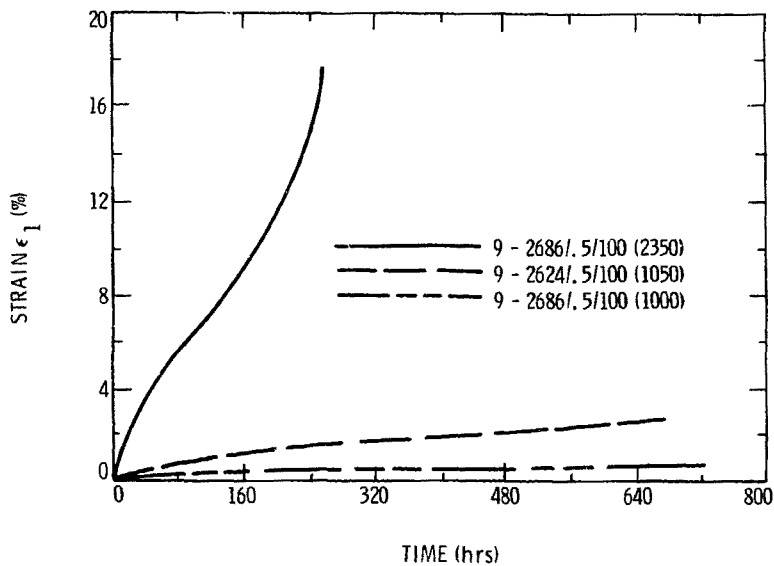


Fig. 10

Axial creep records ( $\epsilon_1$  vs.  $t$ ) at  $\sigma_3 = 500$  psi (3.5 MPa),  
 $T = 100^\circ\text{C}$  and variable principal stress difference (for  
 test code see Fig. 5).



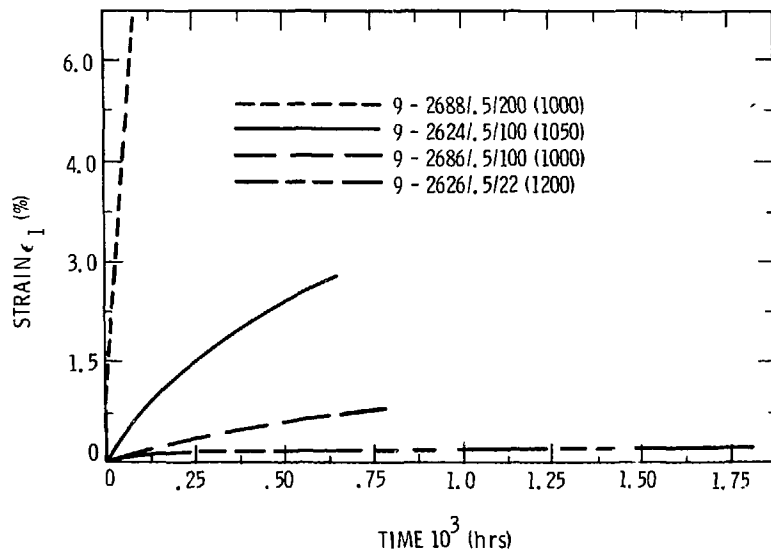


Fig. 11

Axial creep records ( $\epsilon_1$  vs.  $t$ ) at  $\sigma_3 = 500$  psi (3.4 MPa),  
 $(\sigma_1 - \sigma_2) = 1000$  psi (6.9 MPa) and variable temperature  
 (for test code see Fig. 5).

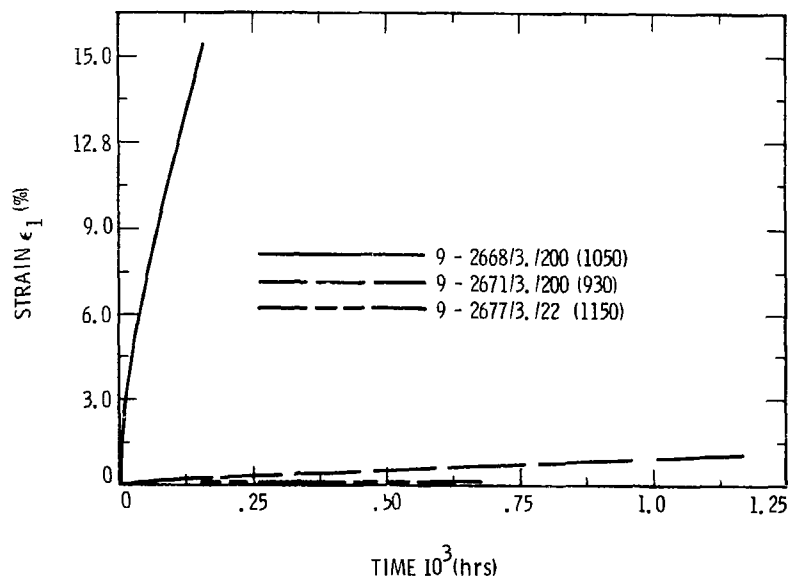


FIG. 17

Axial creep behavior for  $\sigma_1 = 2.0$  psi (13.8 MPa),  
 $(\sigma_2 - \sigma_3) = 1.0$  psi (6.9 MPa) and variable temperature  
 (for test data see Fig. 11).

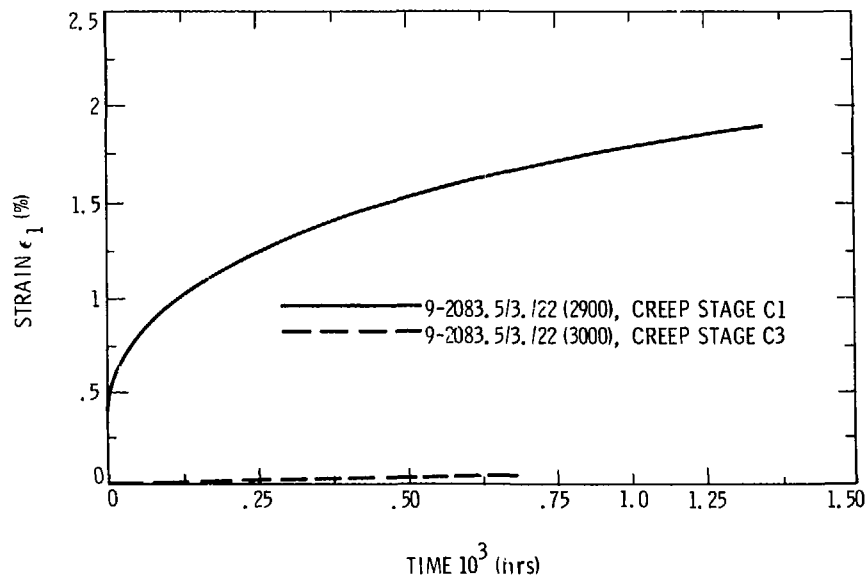


Fig. 23

Axial strain records ( $\epsilon_1$  vs.  $t$ ) for sample 9-2083.5/3.122, creep stage C3, produced at  $(\sigma_1 - \sigma_3) = 4000$  psi (27.1 MPa).

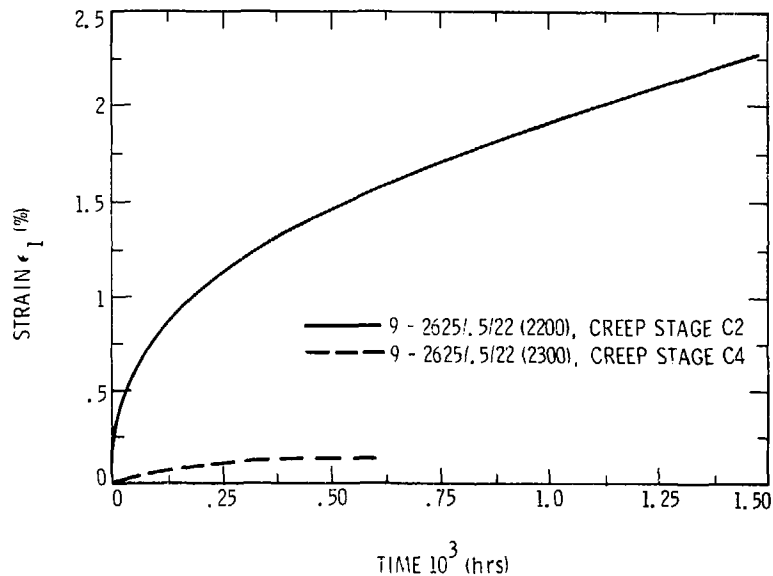


Fig. 1.

Axial strain recorded at wall of Conduits A-B-C, (Inch)  
 Stage C2 provided at  $\sigma_1 = \sigma_2 = 1000$  psi (68.9 MPa).

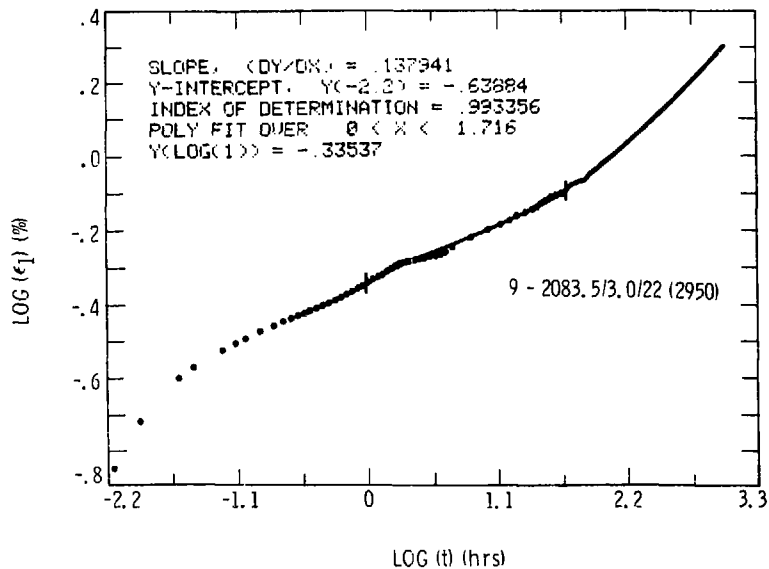


Fig. 17

1. A line is drawn through the data points in Fig. 17. The line is defined by the equation:

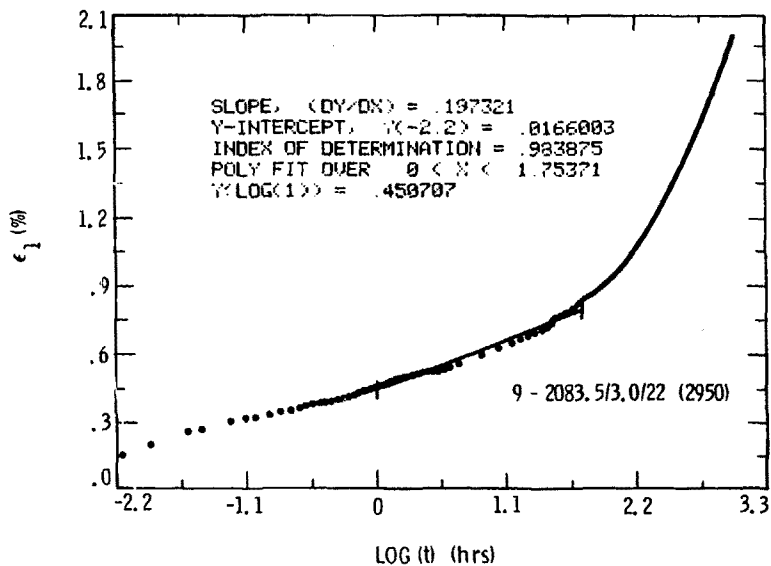


Fig. 16

Semi logarithmic plot  $\epsilon_1$  vs.  $\log(t)$  of test 9-2803/5/22 (2950)  
 (for test code see Fig. 5).

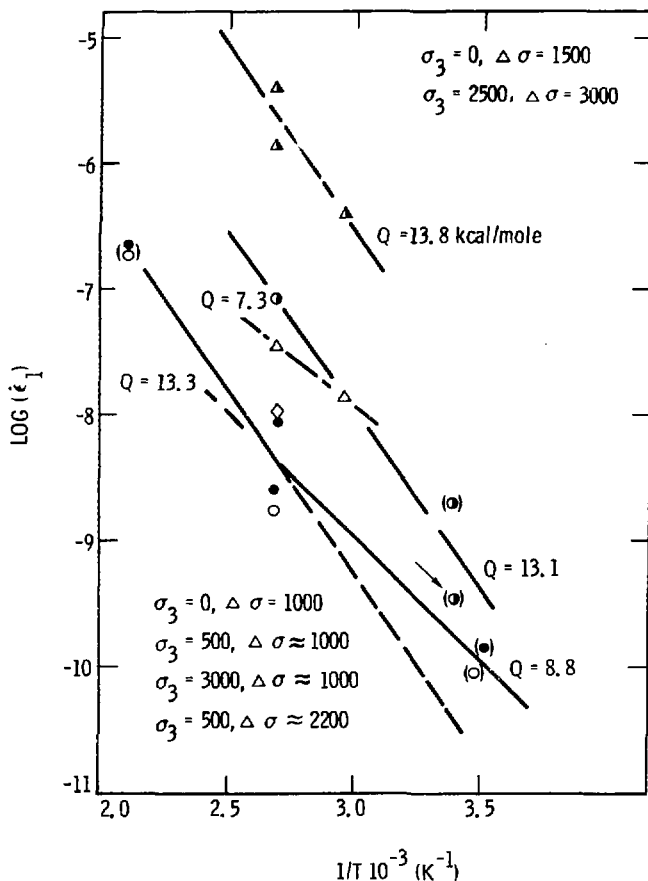


Fig. 17

Secondary creep rates  $\dot{\epsilon}_1$  for lower level salt in the space  $(\log(\dot{\epsilon}_1), 1/T)$ . Triangular data points are due to RE/SPEC, Inc. (Ref. 11).  $\Delta\sigma = (\sigma_2 - \sigma_3)$ . Stresses are given in psi, strain rates in  $(s^{-1})$ . Parentheses denote upper bounds.

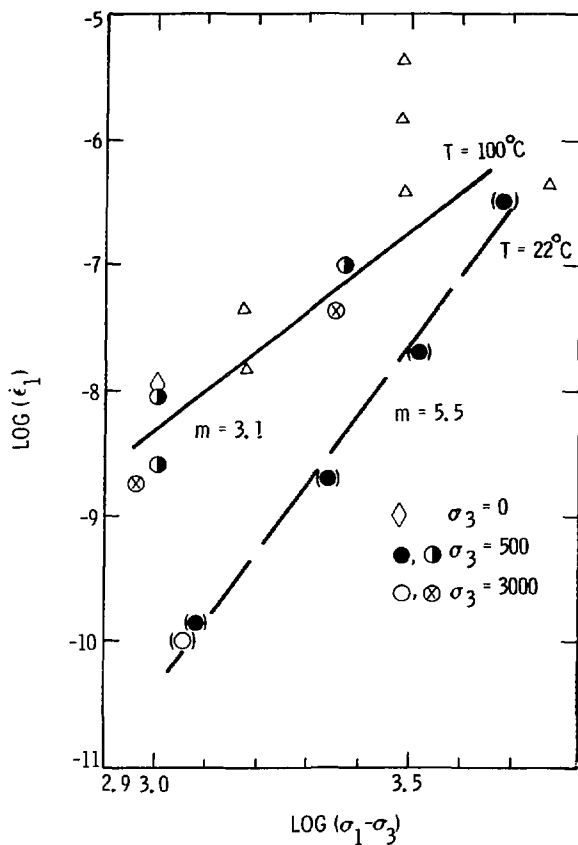


FIG. 18

Secondary creep rates  $\dot{\epsilon}_1$  in space  $\log(\dot{\epsilon}_1)$ ,  $\log(\sigma_1 - \sigma_3)$  for lower level salt. Triangular data points are due to Rt/SPEC, Inc. (Ref. 11). Stresses are given in psi, strain rates in ( $\text{s}^{-1}$ ). Parentheses denote upper bounds.



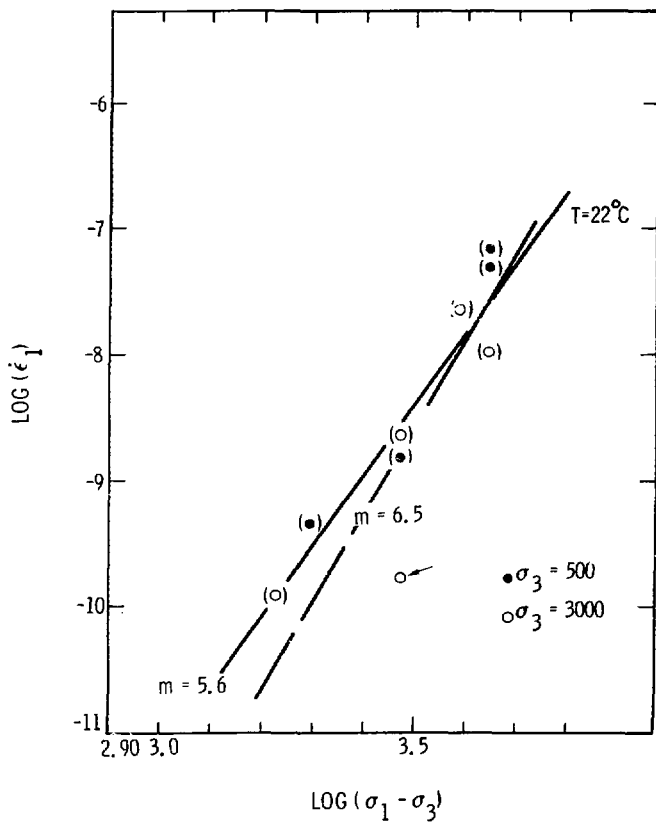


Fig. 10

Secondary creep rates  $\dot{\epsilon}_1$  in the space  $\log(\dot{\epsilon}_1)$ ,  $\log(\sigma_1 - \sigma_3)$  for upper level salt. Stresses are given in psi, strain rate: in  $(s^{-1})$ . Parentheses denote upper bounds.

#### APPENDIX

*Listing of Representative, Condensed Creep Data Files.*

Note abbreviated test code (I.D.): Creep stage no./  
drillhole no. - core depth in ft/containing pressure in  
ksi/test temperature in °F.

TEST I.D. C1/9-2078/.5/22

PRINC. STRESS DIFFERENCE 2000 PSI (13.8 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 3.911         | .152      |
| 8.889         | .164      |
| 20.267        | .174      |
| 24.889        | .187      |
| 192.356       | .25       |
| 227.556       | .256      |
| 261.689       | .262      |
| 284.889       | .267      |
| 310.756       | .272      |

TEST I.D. C2/9-2078/.5/22

PRINC. STRESS DIFFERENCE 4500 PSI (31.0 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 8.2           | 1.822     |
| 24            | 2.63      |
| 33.4          | 3.42      |
| 45            | 3.867     |
| 61            | 4.623     |
| 72.8          | 4.933     |
| 86            | 5.466     |
| 103.8         | 5.981     |
| 125.2         | 6.514     |
| 142.8         | 6.944     |
| 165.2         | 7.477     |

TEST I.D. C1/9-2083/.5/22

PRINC. STRESS DIFFERENCE 2950 PSI (20.3 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 35.778        | .698      |
| 105.778       | .99       |
| 253.556       | 1.241     |
| 412.222       | 1.406     |
| 449.556       | 1.534     |
| 497.778       | 1.605     |
| 595.778       | 1.718     |
| 737.333       | 1.845     |
| 841.556       | 1.95      |
| 942.667       | 2.051     |
| 1078          | 2.19      |
| 1222.67       | 2.34      |

TEST I.D. C2/9-2083/.5/22

PRINC. STRESS DIFFERENCE 4500 PSI (31.0 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 5.778         | 1.242     |
| 24.444        | 2.813     |
| 56            | 3.938     |
| 103.111       | 5.578     |
| 150.222       | 6.68      |
| 210.667       | 7.875     |
| 257.333       | 9.07      |
| 294.667       | 10.43     |
| 331.556       | 11.766    |
| 356.444       | 13.242    |

TEST I.D. C1/9-2070.5/3.0/22

PRINC. STRESS DIFFERENCE 1700 PSI (11.7 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 5.689         | .07       |
| 20.978        | .079      |
| 24.889        | .085      |
| 190.933       | .11       |
| 206.933       | .116      |
| 226.133       | .118      |
| 252.089       | .118      |
| 280.889       | .122      |
| 301.511       | .123      |
| 310.756       | .125      |

TEST I.D. C2/9-2070.5/3.0/22

PRINC. STRESS DIFFERENCE 4100 PSI (28.3 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 12.8          | 1.269     |
| 31            | 1.831     |
| 53.4          | 2.294     |
| 83.2          | 2.738     |
| 102.8         | 2.969     |
| 127.2         | 3.244     |
| 146           | 3.45      |
| 161.8         | 3.613     |

TEST I.D. C1/9-2003.5/3.0/22

PRINC. STRESS DIFFERENCE 2900 PSI (20.0 MPa)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 60            | .85       |
| 146.667       | 1.04      |
| 238.667       | 1.217     |
| 364           | 1.385     |
| 494.667       | 1.509     |
| 624           | 1.644     |
| 766.667       | 1.746     |
| 914.667       | 1.88      |
| 1085.33       | 2.008     |

TEST I.D. C2/9-2003.5/3.0/22

PRINC. STRESS DIFFERENCE 4400 PSI (30.3 MPa)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 26.444        | 2.306     |
| 88.667        | 4.031     |
| 180.444       | 5.344     |
| 309.556       | 6.544     |
| 466.667       | 7.688     |
| 653.333       | 8.756     |
| 857.111       | 9.694     |
| 1067.11       | 10.669    |
| 1247.56       | 11.325    |

TEST I.D. C3/9-2063.5/3.0/22

PRINC. STRESS DIFFERENCE 3000 PSI (20.7 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 6.044         | -.003     |
| 83.867        | .003      |
| 155.644       | .007      |
| 219.111       | .012      |
| 292.4         | .016      |
| 364.178       | .021      |
| 425.378       | .025      |
| 510           | .029      |
| 563.644       | .032      |
| 584.8         | .036      |
| 673.956       | .042      |

TEST I.D. C1/9-2625/.5/22

PRINC. STRESS DIFFERENCE 1200 PSI (8.3 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 102.222       | .131      |
| 360           | .165      |
| 522.222       | .168      |
| 788.889       | .179      |
| 953.333       | .184      |
| 1140          | .197      |
| 1328.89       | .199      |
| 1433.33       | .208      |
| 1648.89       | .213      |
| 1837.78       | .217      |

TEST I.D. C2/9-2625/.5/22

PRINC. STRESS DIFFERENCE 2200 PSI (15.2 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 16            | .424      |
| 83.556        | .716      |
| 176           | .99       |
| 300.444       | 1.193     |
| 451.556       | 1.41      |
| 675.556       | 1.635     |
| 906.667       | 1.834     |
| 1153.78       | 2.033     |
| 1438.22       | 2.246     |

8

TEST I.D. C3/9-2625/.5/22

PRINC. STRESS DIFFERENCE 3300 PSI (22.8 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 10            | 1.033     |
| 31.6          | 1.654     |
| 68.8          | 2.363     |
| 107.2         | 2.888     |
| 156.0         | 3.351     |
| 200           | 3.789     |
| 255.2         | 4.288     |
| 308.4         | 4.673     |
| 355.2         | 5.023     |



TEST I.D. C4/9-2625/.5/22

PRINC. STRESS DIFFERENCE 2300 PSI (16.1 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 22.044        | .04       |
| 71.111        | .054      |
| 137.956       | .075      |
| 209.067       | .089      |
| 303.644       | .103      |
| 393.244       | .114      |
| 469.333       | .119      |
| 504.178       | .12       |
| 544           | .131      |
| 593.778       | .131      |

TEST I.D. C1/9-2677/3.0/22

PRINC. STRESS DIFFERENCE 1150 PSI (8.0 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 29.467        | .093      |
| 102           | .115      |
| 195.689       | .131      |
| 282.578       | .135      |
| 385.333       | .141      |
| 500.178       | .148      |
| 639.956       | .152      |
| 670.933       | .153      |

TEST I.D. C1/9-2672.5/.03/100

PRINC. STRESS DIFFERENCE 1050 PSI (7.3 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 19.2          | .285      |
| 63.289        | .585      |
| 146.489       | .99       |
| 215.467       | 1.37      |
| 310.044       | 1.84      |
| 398.933       | 2.23      |
| 486.4         | 2.62      |
| 603.733       | 3.17      |

TEST I.D. C1/9-2624/.5/100

PRINC. STRESS DIFFERENCE 1000 PSI (6.9 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 18.889        | .319      |
| 44.578        | .512      |
| 78.578        | .731      |
| 127.689       | 1.041     |
| 200.978       | 1.348     |
| 327.156       | 1.772     |
| 399.689       | 2.03      |
| 494.133       | 2.314     |
| 578           | 2.581     |
| 652.8         | 2.783     |

TEST I.D. C1/9-2686/.5/100

PRINC. STRESS DIFFERENCE 1000 PSI (6.9 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 53.2          | .11       |
| 120.756       | .196      |
| 200.133       | .286      |
| 297.244       | .398      |
| 413.778       | .505      |
| 529.467       | .618      |
| 636.711       | .731      |
| 737.2         | .823      |

TEST I.D. C2/9-2686/.5/100

PRINC. STRESS DIFFERENCE 2350 PSI (16.4 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 30.8          | 2.75      |
| 64.4          | 4.781     |
| 95.822        | 6.281     |
| 125.067       | 7.313     |
| 154.933       | 8.75      |
| 184.489       | 10.406    |
| 218.4         | 12.986    |
| 247.333       | 16.063    |
| 260.4         | 17.625    |

TEST I.D. C1/9-2671/3.0/100

PRINC. STRESS DIFFERENCE 930 PSI (6.4 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 69.333        | .116      |
| 268           | .308      |
| 417.333       | .413      |
| 556           | .52       |
| 785.333       | .7        |
| 902.667       | .798      |
| 1065.33       | .909      |
| 1172          | .978      |

TEST I.D. C2/9-2671/3.0/100

PRINC. STRESS DIFFERENCE 2250 PSI (15.9 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 23.467        | 2         |
| 69.422        | 3.55      |
| 127.111       | 5.15      |
| 317.778       | 9.5       |
| 407.733       | 10.75     |
| 536.8         | 13.35     |
| 677.6         | 18.15     |
| 773.422       | 23.1      |
| 866.311       | 29.1      |

TEST I.D. C/9-2688/.5/200

PRINC. STRESS DIFFERENCE 1000 PSI (6.9 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 9.147         | 1.785     |
| 22.12         | 2.731     |
| 29.587        | 3.443     |
| 40.413        | 4.176     |
| 52.08         | 4.856     |
| 64.773        | 5.61      |
| 79.8          | 6.449     |

TEST I.D. C1/9-2668/3.0/200

PRINC. STRESS DIFFERENCE 1000 PSI (6.9 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 13.2          | 2.675     |
| 36.8          | 5.25      |
| 59.8          | 7.525     |
| 83.4          | 9.525     |
| 118           | 12.05     |
| 141.2         | 14.175    |
| 164.8         | 16.05     |

TEST I.D. C/7-2777/.5/22

PRINC STRESS DIFFERENCE 4800 PSI (33.1 MPA)

| TIME<br>(HRS) | E1<br>(%) |
|---------------|-----------|
| 0             | 0         |
| 9.667         | 6.281     |
| 20.222        | 7.25      |
| 29.778        | 8.5       |
| 39.111        | 9.281     |
| 48.667        | 9.813     |
| 55.444        | 12.063    |
| 63.222        | 13.094    |
| 70.444        | 13.625    |
| 80            | 15.469    |
| 89.333        | 16.531    |
| 99.222        | 17.688    |



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