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Properties of Salt-Saturated Concrete and Grout after Six Years In Situ at the Waste Isolation Pilot Plant

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ABSTRACT

Samples of concrete and grout were recovered from short boreholes in the repository floor at the Waste Isolation Pilot Plant more than six years after the concrete and grout were placed. Plugs from the Plug Test Matrix of the Plugging and Sealing Program of Sandia National Laboratories were overcored to include a shell of host rock. The cores were analyzed at the Waterways Experiment Station to assess their condition after six years of service, having potentially been exposed to those aspects of their service environment (salt, brine, fracturing, anhydrite, etc.) that could cause deterioration.

Measured values of compressive strength and pulse velocity of both the grout and the concrete equaled or exceeded values from tests performed on laboratory-tested samples of the same mixtures at ages of one month to one year after casting. The phase assemblages had changed very little. Materials performed as intended and showed virtually no chemical or physical evidence of deterioration. The lowest values for strength and pulse velocity were measured for samples taken from the Disturbed Rock Zone, indicating the influence of cracking in this zone on the properties of enclosed seal materials.

There was evidence of movement of brine in the system. Crystalline phases containing magnesium, potassium, sulfate, and other ions had been deposited on free surfaces in fractures and pilot holes. There was a reaction rim in the anhydrite immediately surrounding each recovered borehole plug, suggesting interaction between grout or concrete and host rock. However, the chemical changes apparent in this reaction rim were not reflected in the chemical composition of the adjacent concrete or grout. The grout and concrete studied here showed no signs of the deterioration found to have occurred in some parts of the concrete liner of the Waste Isolation Pilot Plant waste handling shaft.

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INTRODUCTION

The Waste Isolation Pilot Plant (WIPP) is a research and development facility of the US Department of Energy (DOE). Its purpose is to demonstrate the safe disposal of radioactive wastes resulting from US defense activities. The WIPP repository facility is located in bedded evaporite rock (principally halite), approximately 2,150 ft (656 m) below the ground surface, near Carlsbad, New Mexico.

For over 15 years, the US Army Engineer Waterways Experiment Station (WES), in cooperation with Sandia National Laboratories (SNL), has provided support in research and development of cement-based grouts and concretes for use in the WIPP repository facility. This support focused on the SNL Plugging and Sealing Program (PSP) to develop acceptable sealing technology at the WIPP based on a technical program, including laboratory materials research and verification and field testing (Stormont, 1984). "Sealing" specifically means creating a flow-restrictive barrier to prevent penetrations associated with boreholes, shafts, storage rooms, and accessways from becoming preferred pathways for waste transport after closure.

From 1984 through 1986, SNL performed a series of in situ experiments to evaluate the performance of various candidate seal materials. Field experiments included vertical and horizontal boreholes plugged with salt-saturated grout and concrete as part of the Small-Scale Seal Performance Tests (SSSPT) (Stormont, 1986). The WES formulated these candidate cement-based materials and performed supporting research.

Objectives of the Original Plug Test Matrix

A necessary part of the PSP not addressed by SSSPT was a requirement to assess the stability of candidate seal materials exposed to actual WIPP environments over time. This initiated the Plug Test Matrix (PTM) Test Plan prepared by Gulick (1984) with the purpose of measuring long-term effects on cement-based materials subjected to in situ conditions. Materials placed in boreholes located in the WIPP repository would undergo long-term (five and more years) exposure to physical and chemical conditions of the repository environment. As initially planned, hardened and appropriately aged specimens from the PTM were to be removed at intervals and assessed for physical properties and chemical stability.

Candidate cement-based mixtures formulated for the SSSPT were placed in the WIPP in 1984 and 1985. A saltwater grout (BCT-1F) was developed in 1979 as a result of extensive work by the WES and others in support of the Bell Canyon Test (BCT) for cement grouts (Gulick et al., 1980). Formulation of cementitious material placed in the PTM in late 1984 was essentially the original formulation of BCT-1F grout. Field-cast specimens were prepared and tested at the WES. An expansive salt-saturated concrete (ESC) was also developed at the WES during 1985 (Wakeley and Walley, 1986) and placed in test series A and B of the SSSPT (Stormont, 1986). Field specimens cast during these placements of ESC were studied at the WES, more extensively than were specimens from previous placements. Some of these data were reported previously (Comes et al., 1987). Summaries of grout and concrete development for the WIPP, including WES field and laboratory support to SNL, have also been published (Wakeley and Walley, 1986; Wakeley, 1987; Wakeley and Poole, 1986; Wakeley and Poole, 1987; Gulick and Wakeley, 1989).

As best-candidate mixtures at that time, these materials were emplaced in small-diameter vertical boreholes in the PTM. The PTM consisted of 24 boreholes, each having a 12-ft vertical depth. These boreholes were located in Room L1 of the WIPP repository in two rows parallel to the north wall. The first row consisted of 12 holes, approximately 36 in. apart, each with a 2-in. nominal diameter. The second row, approximately 36 in. behind the first row, consisted of another 12 holes, with a 4-in. nominal diameter. The first row of holes (2 in.) was plugged in 1984 with WES-formulated BCT-1F grout; the second row of holes (4 in.) was plugged in 1985 with WES-formulated ESC during the field activity of placing ESC in the vertical boreholes of the SSSPT. Some of these data were reported previously (Comes et al., 1987).

As described in the original 1984 SNL Test Plan, objectives of the PTM were to:

- Develop emplacement techniques for primary candidate plug materials.
- Provide long-term in situ specimens of candidate plug materials for periodic removal and evaluation of plug properties.
- Expand plug performance data base.
- Verify existing laboratory data from candidate cement-based materials.

Because of changes in priorities, PTM specimens were not recovered or analyzed at the intended intervals. Other than a few portions of plugs recovered in 1985, no samples from the PTM were recovered or analyzed until 1991. By that time, performance criteria for cement-based materials had changed, and BCT-1F and ESC were no longer the leading candidate grout and concrete, respectively.

Discovery of deteriorated concrete in the waste-handling shaft at the WIPP in spring of 1990 led to an extensive research effort to determine the causes of observed deterioration of the concrete of the shaft liner, which was close in age to the SSSPT concrete. Following analysis at the WES and elsewhere of samples of deteriorated concrete recovered from the waste-shaft liner in 1990 (Wakeley et al., 1992; Nowak et al., 1993), it was clear that assessing the condition of concretes that had been in service for roughly the same length of time as the waste-shaft liner would provide data important to understanding the serviceability of concretes in this difficult environment.

The service environments of the waste-shaft liner and the PTM were not directly comparable. Brines in the environments of the waste-shaft liner and the PTM plugs probably were from different sources. There was evident continuous flow of brine at the waste shaft liner. The ratio of potassium to magnesium in seepage brines in the shaft indicated that brines collected there were derived from the contact between the Rustler and Salado Formations (Lambert et al., 1992). Brines available to the PTM were more likely to have been derived from the enclosing rocks at the WIPP storage horizon. The PTM brines were probably more representative of the environment in which concrete-seal components will be placed. Because of this probable difference in source, the chemical compositions and quantity of fluids available to react with the concretes were different, and the materials had not been exposed to identical hazards. However, given the possibility of concrete deterioration in contact with any Mg-bearing fluid, it was important to determine the condition of the PTM plugs.

The original composition and physical properties of the grout and concrete in the PTM had been documented to study any alteration from the original condition. If the PTM grout or concrete were deteriorated to a degree similar of the waste-shaft concrete at the joint between liner and keyway, then significant concerns would obviously exist regarding use of concrete in the seal system. Conversely, if the PTM plugs retained their original composition and properties, they would provide a well documented example of acceptable concrete performance underground at the WIPP.

Objectives of the 1991 PTM Recovery and Analysis

This report documents the recovery and analysis in 1991 of sections of PTM borehole plugs and enclosing rock, approximately six years after placement. This report will address two of the original PTM objectives, with specific emphasis on:

- Assessing indications of long-term performance of original WES-formulated cement-based materials;
- Identifying evidence of any active chemical processes that have affected these materials.;
- Comparing properties of recently recovered PTM materials to those of laboratory- and field-cast specimens of the materials as originally placed and to other reference properties;
- Providing input to decisions made in the ongoing process of refining candidate cement-based materials for the WIPP seal system.

Durability itself is not a measurable property of concrete. Assessing durability is an assessment of performance of a specific concrete in its service environment over time. This being the intent of the PTM, in 1991 four of the PTM borehole plugs were removed and transported to the WES for petrographic analyses and evaluation of physical properties. The following section describes recovery operations. Subsequent sections describe physical testing and assessment of chemical stability.

1991 CORE RECOVERY OPERATION

Two BCT-1F grout plugs and two ESC concrete plugs and their enclosing host rock were overcored between 14 and 23 May 1991 and shipped to the WES for study. This involved a joint effort of SNL, Westinghouse Corp., and other contractors with WES personnel participating between 14 and 17 May. Overcores were taken at twice the actual plugged diameter to preserve the integrity of the host rock and plug interface, and also to minimize the disturbance to the plug. The recovery plan was documented in WES Research Project Plan 91-13(B).

The 12-ft length of grout and concrete plugs extended through three distinct rock zones beneath the repository floor. From the repository floor to depths of approximately 4 to 5 ft, the host-rock material was primarily halite. The surface layer of halite is within the disturbed rock zone (DRZ), defined as the zone of rock in which the mechanical and hydrologic properties have changed in response to excavation (Borns and Stormont, 1989). The anhydrite seam also within the DRZ, called Marker Bed 139 (MB139), lies approximately 5 to 8 ft below the floor surface. Below this depth, halite again was the predominant rock. A layer of up to 2 ft of mixed mineralogy was present at the interface of anhydrite and halite. As documented elsewhere (Borns and Stormont, 1989), the DRZ including MB139 experienced movement and severe fracturing over time. The lower layer of halite was relatively unaffected by movement with significantly less fracturing over time.

The first two holes overcored were from the first PTM row. These were overcored using a 4-in.-inside-diameter core barrel around a 2-in.-diameter BCT-1F grout plug. The first hole, L1S02, was completed by 1500 hrs of day one. Approximately 5 hours of continuous work were required to recover 12 ft of material from this hole. Recovered material had multiple horizontal fractures, with a distance between fractures varying from 0.25 to 3.00 in. Some depths yielded extensively broken core fragments (rubble). This phenomenon was common in the DRZ and MB139 seam. Once drilling passed through the anhydrite seam into the undisturbed salt, the remaining material was recovered in more continuous lengths (approximately 0.5 to 2.0 ft).

Overcoring of L1S02 resulted in recovery of rubble and in some instances lost material. On-site personnel judged that lost material was attributable to jamming of cut material eccentrically in the core barrel, which then would grind the remaining length of the coring run. Although this would have produced large volumes of finely ground rock, it went unnoticed because the vacuum hose system for dust control directly removed all finely ground material from the hole. Overall, core

recovery from L1S02 was not completely successful because of some lost intervals and excessive fracturing.

The second hole overcored was BCT-1F plug hole L1S01, immediately adjacent to L1S02. Again in the DRZ and MB139, recovered material experienced extensive horizontal fracturing. Also, cores were recovered as rubble at certain depths. However, loss of core was minimized because during the cutting procedure, the crew took steps to ensure that the core-barrel rotational speed was kept to a minimum. Also the dust coming from the hole under vacuum pressure was checked continually for evidence of increased volume or changes in particle size.

ESC material was recovered with better results. The same procedures were used as with the BCT-1F plugs, except an 8-in.-inside-diameter core barrel was used to recover 4-in.-diameter plugs. Recovered ESC plug holes were L1S13 and L1S14 (Appendix B). Recovered overcore lengths varied from 6 to approximately 36 in. Minimal rubble material was recovered with L1S14. Although overall recovery of ESC and host-rock material proceeded better than that experienced with grout plugs, other problems occurred that resulted in longer recovery time per hole.

During overcoring of ESC holes, the presence of brine was evident at certain locations. Table 1 tabulates information on hole, wet/dry condition, and rock type. Due to the presence of brine in the borehole, overcoring took longer because the core barrel cutting-teeth experienced a buildup of paste-like material. This was apparently a mixture of brine and powder from cut material. Also, removal of cut material from the borehole was more difficult for these depths. No brine accumulations had been noted during recovery of the BCT-1F grout plugs, so it was not possible to recover brine from holes drilled during this operation. Grout holes were dry during recovery and were observed by WES personnel to have remained dry 48 hours thereafter. Open holes nearby were at least partly filled with fluid, which onsite personnel attributed to human activity, including brine sprayed for dust control.

Table 1. Expansive ESC Samples Associated with Brine on Recovery

<u>Recovered Sample</u>	<u>Hole Depth</u>	<u>Condition</u>	<u>Rock Zone</u>
L1S13 - No.1	0.58 to 2.88 ft.	Wet	DRZ
L1S13 - No.2	2.88 to 5.38 ft.	Wet	DRZ/MB139
L1S13 - No.5	9.92 to 11.67 ft.	Wet	Halite below MB139
L1S14 - No.2	2.25 to 4.25 ft.	Wet	DRZ

Repository room temperatures, relative humidity, and core temperatures were measured during overcoring and core recovery. Room temperatures ranged from 80 to 83°F, and relative humidity measurements ranged from 18 to 31 percent. Immediately after removal from boreholes, core temperatures measured approximately 90°F. Based on core surface temperature measurements after removal from holes, core temperatures during barrel cutting was estimated to have been $\geq 100^{\circ}\text{F}$. This was a concern because of the potential for heat to alter the phase composition of cement-based materials.

Recovered materials were then prepared and shipped to the WES under the supervision of SNL representatives at the WIPP site. Materials were wrapped in plastic then boxed and crated while underground at the WIPP facility. SNL quality assurance forms were used to document shipping and receiving. Overcoring was completed on 23 May, and materials were received at the WES on 6 June 1991.

Materials received at the WES arrived in poor condition because they were inadequately packaged at the WIPP site before shipping. The plastic film used to wrap core segments and intended to seal them from each other and from environmental exposure was open and torn. This caused samples to be exposed to unknown conditions during shipping and allowed physical mixing of some materials where rubble had been recovered. Some core segments had been broken; all core fragments of questionable identity were discarded. This limited our ability to reconstruct core lengths and conduct initial visual examinations and decreased the number of samples available for testing.

PHYSICAL CONDITION AND PROPERTIES

All core samples recovered in 1991 were examined visually in the as-received condition. Condition of cores and fracture surfaces was noted. Specimens of each core were identified for specific tests and analyses based on the relative length of the sample and its physical condition. Specimens selected were from the surface, DRZ, MB139, and from material representing wet and dry conditions when recovered. Tests selected for assessing physical properties are shown in Table 2, with core intervals assigned to each test type. Materials were also scheduled for petrographic examination. The purpose of all analyses was to assess the long-term stability of these materials after extended exposure to WIPP repository conditions.

Table 2. Schedule for PTM Recovered Materials

<u>Sample</u>	<u>Physical Property Analyses</u>		
	<u>Pulse Velocity</u>	<u>Compressive Strength</u>	<u>Push-Out (Shear)</u>
<u>BCT-1F:</u>			
<u>L1S01 - No. 6</u>			
8.17 to 9.08 ft.	✓		✓
9.17 to 9.83 ft.	✓	✓	
9.83 to 10.58 ft.	✓		✓
<u>L1S02 - No. 1</u>			
0.00 to 0.33 ft.	✓		✓
1.00 to 1.25 ft.	✓	✓	
1.58 to 1.83 ft.	✓		✓
<u>L1S02 - No. 3</u>			
5.42 to 5.67 ft.	✓	✓	
6.67 to 7.17 ft.	✓		✓
<u>ESC:</u>			
<u>L1S13 - No. 1</u>			
0.58 to 1.83 ft.	✓		✓
1.83 to 2.88 ft.	✓	✓	
<u>L1S13 - No. 3</u>			
6.38 to 7.50 ft.	✓	✓	
<u>L1S13 - No. 4</u>			
7.67 to 8.50 ft.	✓		✓
8.58 to 9.92 ft.	✓	✓	
<u>L1S14 - No. 2</u>			
2.25 to 2.88 ft.	✓		✓
2.88 to 3.50 ft.	✓	✓	
<u>L1S14 - No. 4</u>			
6.50 to 7.42 ft.	✓		✓
<u>L1S14 - No. 5</u>			
7.83 to 9.00 ft.	✓	✓	

PHYSICAL PROPERTY TESTING

Samples outlined in Table 1 were prepared for pulse velocity, compressive strength, and push-out testing. Pulse velocity and compressive strength testing followed standard American Society for Testing and Materials (ASTM) procedures C 597-83 *Standard Test Method for Pulse Velocity Through Concrete* and C 39-86 *Standard Test Method for Compressive Strength of Cylindrical Concrete Specimens* (American Society for Testing and Materials, 1991). These tests were conducted to assess the relative quality of cement-based specimens as compared to similar data generated from specimens cast during original placements.

Placement of BCT-1F material occurred in 1984 with testing of field-cast specimens in 1984 and 1987. These data from tests of BCT-1F have not been published before, although the purpose of the field activities and some test data have been published (Wakeley, 1987; Gulick and Wakeley, 1989). Specimens were cast in the field in both plastic and steel molds, representing unrestrained and restrained conditions, respectively. Original mixture proportions for this grout are in Appendix A.

Placement of ESC occurred in 1985 with testing of field-cast specimens for compressive strength and other physical properties occurring periodically at the WES for up to one year. Some of these data were analyzed and published previously (Comes et al., 1987). Pulse velocity measurements were not conducted as part of that ESC test program. Mixture proportions for the field-cast ESC as reported by Wakeley and Walley (1986) are in Appendix A.

TESTS OF CORES RECOVERED IN 1991

Pulse velocity tests were conducted on each sample identified for physical property testing (Table 2). Data for dynamic modulus of elasticity exist for both ESC and BCT-1F field-cast specimens. But this property was not measured on recovered cores because individual samples were too small. Data from individual tests of pulse velocity and other properties for recovered cores are tabulated in Appendix B. Data for specimens cast during original placements of BCT-1F are tabulated in Appendix A.

Data from tests of pulse velocity for BCT-1F grout, as measured for the plugs recovered in 1991, are compared to data from similar tests of grout specimens cast at the same time and tested in 1984 (initial) and 1987 (Figure 1). Comparison of current ESC data with pulse velocity values published for concrete (Malhotra, 1976) is shown in Figure 2. Specimens were prepared for tests of compressive strength from both ESC and BCT-1F cores recovered in 1991. ESC specimens were prepared by removing 2-by-4-in. concrete cores from the center of the original 8-in.-diameter recovered cores. Samples L1S14 - No.2 (2.88 to 3.50 ft.) and - No.5 (7.83 to 9.00 ft.) were not tested as planned because problems occurred while attempting to retrieve inner cores. Inner cores were not necessary in preparation of BCT-1F specimens for tests of compressive strength. Measured strength values of specimens with length-to-diameter (L/D) ratios less than 1.8 were corrected as indicated in ASTM C 42. Data from these tests are presented in Appendix C.

Data from tests of field-cast ESC cylinders were published previously (Comes et al., 1987) and are not repeated here. Data from tests of BCT-1F field-cast cylinders are presented in Appendix A. Comparison of strength data for BCT-1F is shown in Figure 3. Values from 1984 are from field-cast samples at 104 days of age. The lower values from 1987 were from samples cast and stored in steel molds, many of which fractured when the molds were removed. Higher values from 1987 tests were from samples cast and stored in plastic molds. These experienced less stress relief when molds were removed, resulting in less fracturing and higher measured strengths. Strength values from 1991 tests are higher than either average value from 1987. In Figure 4, data from tests of field-cast samples of ESC are compared to data from PTM samples recovered and tested in 1991. Again, 1991 values from samples that had spent six years in the underground at the WIPP are higher.

Figure 1. Comparison of BCT-1F pulse-velocity data.

Figure 2. Comparison of ESC pulse-velocity data with Malhotra (1976) reference values.

Figure 3. BCT-1F compressive strength data.

Figure 4. ESC compressive strength data.

Push-out tests were conducted to help assess the performance of each cement-based material in the vicinity of the host-rock interface. Previous similar push-out tests were conducted by others (Roy et al., 1985; Akgun and Daemen, 1991). There are uncontrollable factors associated with the push-out testing, such as eccentricity of the plug segment in the total core, an unavoidable condition in overcoring operations. Therefore, the test does not necessarily measure interface bond strength. In cases where the plug supposedly being pushed out is not parallel to the overcore, the supposed "bond strength" may actually be a measure of the strength of the rock.

Another factor that makes bond-strength values inconsistent includes microfractures induced at and around the interface during core recovery. However, if a cement-based material is going to undergo chemical attack, it is most likely to do so at the interface of plug and host rock. So the presence of a secure bond at these interfaces is a good indicator of the concrete performance. When the samples were received at the WES, initial observations of interfaces showed no evidence of plug/rock interaction for samples in rock salt, although there was an apparent reaction on rim in anhydrite. Physical descriptions of these interfaces are presented elsewhere.

The fact that all plug/rock interfaces remained intact during coring, core recovery, shipping, and subsequent sawing and subsampling shows the generally excellent condition of the plug/rock interfaces and favorable performance and durability of these materials after more than six years of exposure to the WIPP environment. Appendix B presents test results for both BCT-1F and ESC plugs; Appendix C provides a description of the test method used for push-out testing. Figure 5 summarizes push-out data on specimens recovered from the WIPP.

Figure 5. Push-out data for specimens tested in 1991.

MICROSTRUCTURE AND PHASE ANALYSIS

The comparison of data from earlier test results on physical properties of the grout and concrete to those generated during the recent study was straightforward. For data from phase composition and microstructure, it was difficult to compare old data—those from samples cast in molds at the time of PTM placements or recovered soon after—to data from samples recovered from the WIPP site during 1991. The technology of analyses had changed. Both the old and new data represent many variables that can change some aspect of microstructure without having a gross impact on physical properties. The stable phase assemblage present in a hydrated cement can be changed notably by temperature and humidity. Old samples had been stored at various temperatures and humidities, none of which was the same as that of the repository. If differences were observed, the source might not be obvious. Compressive strengths increased from initial conditions instead of decreasing, indicating that if chemical changes had occurred, they had not reduced the strength of the grout or concrete.

There is some uncertainty about the variability of the environment into which the PTM plugs were placed. Brines or other fluids could have been introduced during in situ permeability tests or spraying for dust control. Such fluids introduced to the DRZ could have affected a large volume of rock and possibly the test plugs. However, these events occurred five or more years ago. Although it is likely that the plugs could have been exposed to brines of an unknown composition early in their history, it is also likely that brines present now and for some time in the past represent the natural brines of the enclosing strata.

As stated earlier, only four intervals of concrete core were visibly wet when recovered, and none of the holes accumulated a noticeable amount of fluid in the few days after the 1991 drilling. Three of the wet intervals were in the DRZ, and the fourth was in the halite below the marker bed. However, virtually all concrete core segments were wet during their time in situ. Evidence for this is the presence of crystalline material in cracks, pilot holes (Figure 6), and all free surfaces, consisting of several minerals related chemically to the host rocks.

None of the intervals of grout core was visibly wet when recovered, and there were no open holes within the plugs. However, crystalline compounds had accumulated on surfaces of fractures through the grout and associated with fractures or distinct crystalline stringers in the enclosing rocks. As in the concrete, the mineral accumulations in all intervals of grout were Mg-bearing compounds, such as polyhalite, magnesite, and sulfates. The minerals in certain intervals are listed in the

Figure 6. Gypsum crystals growing out of pilot holes in concrete.

discussions of phase composition and microstructure in the following sections. Magnesite as an accessory mineral to the anhydrite of MB139 is shown in Figure 7, demonstrating the presence of magnesium in the host rock.

As described previously, the recovered core segments represent three different rock zones with different mineral composition and extent of fracturing. Because some intervals of concrete core were wet when recovered, it was possible to compare wet versus dry intervals. Locations of original holes and of recovered plugs were chosen for physical convenience rather than systematically to represent wet or dry conditions at the time the holes were plugged. There also are two materials (grout and concrete) with different porosities, densities, aggregates, and composition of cementitious materials. The number of samples representing any one set of conditions was small, precluding any statistical analyses of data to delineate controlling variables.

We compared phase assemblages and physical appearance of grout or concrete samples by pairs, emphasizing comparisons of wet versus dry cores and of those from the DRZ versus an undisturbed condition. The following paragraphs describe these comparison pairs based on observations from petrographic studies including phase composition and microstructure. Core intervals are given for each pair of cores cited as an example of each condition by field-core number, WES core number, and depth.

Grout Versus Concrete

Previous work on the composition of grout samples (BCT-1F) by x-ray diffraction (XRD) analysis determined that the grout characteristically contained calcium chloro-aluminate hydrate $[\text{Ca}_2\text{Al}(\text{OH})_6] [\text{Cl} \cdot 2\text{H}_2\text{O}]$, ettringite $\{\text{Ca}_6 [\text{Al}(\text{OH})_6]_2 \cdot 24\text{H}_2\text{O}\} [(\text{SO}_4)_3 \cdot 1-1/2\text{H}_2\text{O}]$, and portland-cement phases for ages from 48 days to 32 months after casting. Phase composition of concrete (ESC) was not followed closely after field placement because the material had not been formulated for geochemical stability, and the need to follow chemical changes was not a recognized part of the work under way at the time. Given its component materials, we surmise that its composition was similar to the grout samples.

Figure 7. Scanning electron secondary emission (left) and backscattered emission (right) images of a polished surface of anhydrite.

The grout and concrete taken from the 1991 cores were analyzed by XRD. The only compositional differences identified were attributable to the different aggregates used in the grout (primarily quartz sand) compared to that of the concrete (dolomite, calcite, and quartz). No major differences in phase composition were observed for grout samples taken at different depth intervals or from different host rocks. The phase composition of all grout samples taken from cores was the same as that determined previously, except that ettringite was absent in the 6-year-old grout and anhydrite was observed in samples recovered from the anhydrite zone.

As with the grout samples, all XRD patterns of 1991 concrete have the same phases and in similar proportions, regardless of depth interval. These phases include dolomite [$\text{CaMg}(\text{CO}_3)_2$], calcite [CaCO_3], quartz, ettringite, calcium chloro-aluminate hydrate, and halite.

Wet Versus Dry

Sections of concrete core samples identified as recovered from wet zones all had the same phases and in similar proportions, as determined by XRD. As noted above, these phases included dolomite, calcite, quartz, ettringite, calcium chloro-aluminate hydrate, and halite. Furthermore, the phases present in these concretes at the contact with halite are the same as those in the center of each core, away from the interface or other discontinuities. This indicates that there is no influence on phase composition of the concrete by contact with salt and an apparently stable phase assemblage.

Phases observed in pilot holes in the concrete recovered wet included halite, brucite [$\text{Mg}(\text{OH})_2$], and sylvite [KCl]. There was a large proportion of an apparently amorphous phase present in these locations. Other phases found were well formed gypsum [$\text{CaSO}_4 \cdot 2\text{H}_2\text{O}$] crystals extending out from the wall of the hole into the void (Figure 6).

Concrete samples recovered in dry salt were virtually identical to those recovered wet. The phase assemblages close to the salt contact were indistinguishable by XRD from those in the centers of the cores. Concrete samples recovered in contact with anhydrite included the same phases found in the concretes recovered both wet and dry, as determined by XRD.

Crystallization of various minerals, such as gypsum, also occurred in the concrete samples recovered in a dry condition. The host rock contains the phases found in the fractures in the

concrete. In addition to halite and gypsum, other phases present in the halite host rock included calcite, dolomite, anhydrite, and quartz. An XRD of a seam in the host halite revealed the presence of polyhalite [$K_2Ca_2Mg(SO_4)_4 \cdot 2H_2O$], magnesite [$MgCO_3$], anhydrite [$CaSO_4$], and quartz. Similar phases were observed in longitudinal fractures found in the anhydrite zone.

The presence of Mg-bearing accessory minerals was confirmed in the anhydrite host-rock images from secondary-emission and backscattered-emission scanning electron microscopy (SEM) of a polished surface of anhydrite show that the Mg was present in discrete areas, appearing as dark patches in Figure 7. Ion mapping of the sample indicated that the elements Ca and S were present together (i.e., $CaSO_4$) in the lighter-colored areas, where Mg was absent. The presence of carbon associated with Mg, as determined by energy-dispersive x-ray fluorescence, suggests that $MgCO_3$ is present, possibly with MgO or $Mg(OH)_2$ or both.

The Mg-rich areas in the anhydrite cannot be related to Mg in the grout, because the Mg content of the grout (1.5% by mass) is lower than that of the bulk anhydrite samples (2 to 3% by mass). Bulk analyses also show that there is an increased level of Mg in the rock near the interface compared to the Mg level elsewhere in the rock (<0.5% by mass). By interaction with an Mg-rich drilling fluid or dust-control brine, the Mg level in the vicinity of the hole may have increased after the hole was drilled and before the grout or concrete was poured. There are other possible explanations, but none seems to involve the grout.

The presence of Mg-bearing magnesite and sulfate-bearing gypsum filling cracks in both the concrete and salt layers indicates that these ions are moving in both obviously wet and apparently dry rock zones. "Dry" and "wet" refer to the condition at drilling in 1991. The source could be local or introduced brine and may involve dissolution of pre-existing phases. Furthermore, the presence of brucite only in the pilot holes open in concrete indicated that this phase probably precipitated in situ only where a magnesium source (presumably brine) encountered the increased pH of direct contact with the concrete. There is no evidence of precipitation of any phases directly removed from the concrete and no evidence of the porous microstructure that results from partial dissolution of cement.

One sample studied by SEM revealed perfectly formed clusters of magnesium hydroxy-chloride hydrate [$Mg_2(OH)_3Cl \cdot 4H_2O$] at the interface of host salt and concrete. This phase was common in highly altered portions of the waste-shaft concrete (Wakeley et al., 1992), but its presence was not confirmed by XRD in the PTM samples, so if present it is rare. No Mg-bearing phases are present in

any interval within the matrix of the concrete: they are present only on free surfaces such as cracks and pilot holes, where fluids were free to migrate and where the hydrated cement phases controlled the pH. Although Mg is present in the system, it has not caused the plug materials to alter.

Disturbed Versus Undisturbed Rock

The phase assemblages in concretes recovered from both the disturbed and undisturbed rock zones were the same. Visual examination of the cores revealed large discontinuities spaced at 3- to 5-in. intervals in both the salt and concrete sample cores from the DRZ, usually in association with a discontinuity in the rock such as a clay seam. These discontinuities were mineral-filled fractures. They were observed most easily on intact cores after push-out tests (Figure 8) and on core segments that had been sawn in half longitudinally. As noted previously, fractures also were observed in the concrete in the undisturbed rock, although they were more widely spaced. Even with this obvious difference in fracture frequency and therefore differences in opportunity for interaction with brine, the microstructure and phase composition in these intervals are very similar.

Strong Versus Weak Interface Bond

In general, bonds between salt host rock and concrete appeared strong and intact upon visual examination. Examples of concrete-to-salt bonds show good delineation between the concrete and the salt and look as if the concrete is encased in plastic (Figure 9). Further evidence of the bond integrity between the salt and the concrete is shown in the sample cores after push-out tests were performed (Figure 10). In this case, the failure occurred in the salt, apparently because of the extensive and typical fracture network. In sections of core in the undisturbed halite zone, failure occurred in the concrete, and a visible ring of concrete remained bound to the halite host (Figure 11). These bonds are characteristically stronger than the halite host rock or the concrete and demonstrate the compatibility between these two materials.

Both concrete core segments recovered from MB 139 showed marked reaction rims in the host anhydrite and had formed a weaker bond with the concrete than was exhibited by the lower undisturbed salt (Figure 12). These reaction rims are darker colored and more porous than the host rock. Samples taken from the intersection of the reaction rims and fractures were analyzed by XRD,

Figure 8. The concrete plug from the DRZ after push-out test, showing mineral-filled fractures.

Figure 9. Example of concrete-to-salt bond.

Figure 10. Concrete/halite core after push-out test, with breakthrough halite indicating strong bond between the salt and the concrete.

Figure 11. Concrete/halite core from the undisturbed halite zone where failure occurred in the concrete.

revealing halite, anhydrite, and quartz. Calcite, dolomite, gypsum, and magnesite were minor constituents. Halite was not detected in most reaction rim samples. This may indicate increased dissolution of the more soluble halite compared to the less soluble anhydrite. Further evidence for the poorer bond between the concrete and the anhydrite is shown in the sample cores after push-out tests were performed (Figure 13). In this example, the failure occurred at the interface of the concrete and the anhydrite host rock.

The grout formed a good bond with both the NaCl and CaSO₄ host rock. A similar reaction ring was observed in the anhydrite host rock around both grout cores. The higher measured values for push-out strength of grout cores indicate that eccentricity of the plug in the overcore was more of a problem for the smaller-diameter cores. The very high value reported for push-out strength of grout in MB 139 (Figure 5) shows that the core was the most off-center, the diameter was small, and the anhydrite was stronger than halite. The push-out tests of grout from anhydrite measured the strength of the host rock, giving the highest values measured for this type of testing. The samples also failed along the bond when tested; that is, the rock and the bond failed, but the grout did not.

Figure 13 Concrete/anhidrite core after push-out test. Note that failure occurred in the anhydrite host.

DISCUSSION AND CONCLUSIONS

All tests performed to assess material quality indicated that both the grout and the ESC performed very well. Values for compressive strength and pulse velocity of the grout equaled those measured within a few months after casting. Comparable values for the concrete exceeded previously reported values or exceeded reference values for good concrete. Both materials had visibly good bonds with the host salt rock, with measured bond strength increasing downward from the DRZ to the anhydrite into the undisturbed salt.

Measured push-out strengths for the grout were artificially high because of the eccentricity of samples. Direct observation of samples confirmed the eccentricity and confirmed that the anhydrite host rock had developed a porous reaction rim at the interface; the halite did not. This is consistent with previous studies of simulated borehole plugs in anhydrite in which porous reaction rims developed and prevented formation of a measurable bond (Moore et al., 1980; Gulick et al., 1980; Burkes and Rhoderick, 1983; Buck, 1985). The surprising feature of the present study is that the bond that formed between concrete or grout and anhydrite held the sample together through all cutting and handling.

There is evidence that indicates dissolution of NaCl in the anhydrite zone in the vicinity of the plugs, probably causing the increased porosity in anhydrite reaction rims. There also is evidence that the presence of halite promoted better bonding with the grout and concrete because the few areas that had mixed halite/anhydrite host rock were well bonded to the plugs in the halite-contact regions. There were no major differences in phase assemblage between samples recovered either wet or dry. All showed evidence of crystallization of new phases on free surfaces but no major changes in phase assemblage since the assemblage was originally documented. This was true even where the rock was highly fractured, which could have enhanced brine movement; and where pilot holes were present through the concrete, which could have had the same effect. In fact, the crystallization of new phases on these free surfaces indicates that strongly ionic Mg-bearing fluids were present and moving. But the stability of the phase assemblage in the cement-based materials indicates that they were essentially unaffected by this movement other than to improve bonding to host salt rock.

One of the questions asked about grout or concrete since the inception of geochemical studies in support of the SSSPT is: is the phase assemblage of this material inherently unstable in its service environment? This instability presumably would be indicated by major phase changes in a fairly short

time, with concomitant deterioration in physical properties. It appears that these materials were stable in their service environment, and their physical properties were favored.

The ESC is in much better condition than the concrete taken from the waste shaft at the 834-ft construction joint (Nowak et al., 1993) after roughly the same length of time in service. The concrete in the PTM apparently was exposed to far less fluid and slower fluid transport. It probably also encountered far less magnesium, even though there are Mg-bearing phases even on free surfaces. Mg phases are absent from the hydrated-cement phase assemblage. There is no doubt that in this particular WIPP sub-environment, both the grout and the WES concrete performed well and that long-term chemical stability is indicated as long as service conditions remain constant. That is, where Mg level and fluid availability are low and there is no potential gradient to force the fluids through the concrete, a concrete formulated carefully for this environment should be durable. We cannot judge the relative resistance to Mg-related deterioration of the waste-shaft concrete versus the ESC because they were in such apparently different service conditions. The differences in their appearances, strength, and phase assemblages is striking.

The Mg-bearing phases had not been identified in previous studies of the grout or concrete, from samples cast in the field and stored in air or from small samples recovered in 1985 from the PTM. The fact that Mg phases are present in the 1991 recovered samples, even though they only fill fractures and cling to free surfaces and possibly interfaces, means that Mg was available from some source in this environment and it was mobile. Previous and current studies at the WES and other field experiences show the potential for concrete to deteriorate in high-Mg brines. Despite the obvious availability of Mg, a transport medium, and access of these to the concrete, the concrete and grout were stronger than they were six years previously, and their phase assemblages apparently were stable.

In all conditions and for both materials, the lowest measured values for a particular property were from grout or concrete recovered from the DRZ. The DRZ will present the greatest challenges to material performance and will need special grouting or other treatment to achieve performance compatible with other rock zones in the repository.

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APPENDIX A: MIXTURE PROPORTIONS FOR GROUT AND CONCRETE AND GROUT PROPERTIES MEASURED DURING PREVIOUS RESEARCH

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Table A-1. Components and Proportions of Salt-Saturated Grout (BCT-1F) (from Gulick and Wakeley, 1989)

<u>Component</u>	<u>% of Total by Mass</u>	<u>% of Total Solids by Mass</u>	<u>Batch Weight for 1 ft³, lb</u>
Class H Cement	48.3	61.2	61.9
Class C Fly Ash	16.2	20.6	20.8
Cal Seal (Plaster)	5.7	7.2	7.3
Salt (NaCl)	7.9	10.0	10.1
Dispersant	0.78	1.0	1.0
Defoamer	0.02	0.02	0.02
Water	<u>21.1</u>	<u>-</u>	<u>27.1</u>
	100.0	100.02	128.22

Table A-2. Components and Proportions of Expansive Salt-Saturated Concrete (ESC) (from Wakeley, 1987)

<u>Component</u>	<u>% of Total by Mass</u>	<u>% of Total Solids by Mass</u>	<u>Batch Weight for 1 yd³, lb</u>
Class H Cement	9.03	9.66	365
Chem Comp III	6.02	6.45	243
Cal-Seal	1.80	1.94	73
Class C Fly Ash	5.10	5.44	205
Fine Aggregate	34.11	36.50	1,365
Coarse Aggregate	34.58	37.00	1,381
NaCl	2.50	2.65	100
Defoaming Agent	0.21	0.24	9
Na Citrate	0.11	0.12	4.4
Water (Iced)	<u>6.60</u>	<u>-</u>	<u>292</u>
	100.06	100.00	4,037.4

Table A-3. PTM (BCT-1F) Specimens Tested in 1987

Specimen Size (inches)	Type Mold	Condition after Stripping	Pulse Velocity (ft/sec)	Compressive Strength (psi)
3x6	Steel	Cracked	12,280	4,910
			9,640	2,220
			10,962	3,890
			9,640	3,620
			11,259	4,130
4x8	Plastic		11,726	5,170
			11,549	3,580
			12,121	7,250
			12,287	6,630
			12,333	5,490
			12,191	4,610
			11,607	3,580
6x12	Plastic		11,666	5,310
			12,229	6,540
			12,229	6,970
			11,991	2,190
			11,646	2,920
6x12	Steel		11,173	3,360
			11,048	4,230
			12,339	5,010
			12,229	4,160

Table A-4. PTM (BCT-1F) Specimens Tested in 1984 Cast in Steel Molds

Specimen Size (inches)	Age Days	Compressive Strength (psi)	Pulse Velocity (ft/sec)
2x2x2	30	5,330 5,830 3,520 4,810	
2x2x2	68	5,875 6,500 4,820 5,480	
3x6	103	8,810 8,240	
1x1x12	30		9,417 9,417 9,232 9,061 9,014 9,282 9,191 8,844 8,968 9,014 8,928
2x2x12	30		8,849 9,592 9,566 10,190 9,135
3x3x12	30		8,987 9,232 9,150 9,083

APPENDIX B: DATA FROM TESTS OF PHYSICAL PROPERTIES 1991

Table B-1. Data from Static Pulse Velocity Measurements on BCT-1F Grout Samples Removed from the PTM, 1991

Sample	Pulse Velocity Measurement (ft/sec)	Host Rock
<u>L1S01 - No. 6</u> 8.17 to 9.08 ft.	10,205	Undisturbed Halite
9.17 to 9.83 ft.	9,012	
9.83 to 10.58 ft.	10,417	
<u>L1S02 - No. 1</u>		Disturbed Halite
0.00 to 0.33 ft.	4,412	
1.00 to 1.25 ft.	3,788	
1.58 to 1.83 ft.	5,859	
<u>L1S02 - No. 3</u>		Anhydrite Bed
5.42 to 5.67 ft.	9,167	
6.67 to 7.17 ft.	3,580	

Table B-2. Data from Compression Strength Testing on BCT-1F Grout Samples Removed from the PTM, 1991

Sample	Compressive Strength (lbs/in²)	Host Rock
<u>L1S01 - No. 6</u> 9.17 to 9.83 ft.	8,820	Undisturbed Halite
<u>L1S02 - No. 1</u> 1.00 to 1.25 ft.	5,490	Disturbed Halite
<u>L1S02 - No. 3</u> 5.42 to 5.67 ft.	8,400	Anhydrite

Table B-3. Data from Push-Out Testing on BCT-1F Grout Samples Removed from the PTM, 1991

Sample	Push-Out Strength (lbs/in² Shear)	Host Rock
<u>L1S01 - No. 6</u> 8.17 to 9.08 ft.	488	Undisturbed Halite
9.83 to 10.58 ft.	533	Undisturbed Halite
<u>L1S02 - No. 1</u> 0.00 to 0.33 ft.	149	Disturbed Halite
1.58 to 1.83 ft.	409	Disturbed Halite
<u>L1S02 - No. 3</u> 6.67 to 7.17 ft.	576	Anhydrite

Table B-4. Data from Static Pulse Velocity Measurements on ESC Samples Removed from the PTM, 1991

Sample	Pulse Velocity Measurement (ft/sec)	Host Rock
<u>L1S13 - No. 1</u> 0.58 to 1.83 ft.	7,803	Disturbed Halite
1.83 to 2.88 ft.	14,244	Disturbed Halite
<u>L1S13 - No. 3</u> 6.38 to 7.50 ft.	14,661	Anhydrite Bed
<u>L1S13 - No. 4</u> 7.67 to 8.50 ft.	15,424	Undisturbed Halite
8.58 to 9.92 ft.	15,278	Undisturbed Halite
<u>L1S14 - No. 2</u> 2.25 to 2.88 ft.	15,000	Disturbed Halite
2.88 to 3.50 ft.	14,571	Disturbed Halite
<u>L1S14 - No. 4</u> 6.50 to 7.42 ft.	15,458	Anhydrite Bed
<u>L1S14 - No. 5</u> 7.83 to 9.00 ft.	14,524	Undisturbed Halite

Table B-5. Data from Compressive Strength Testing on ESC Samples Removed from the PTM, 1991

Sample	Compressive Strength (lbs/in²)	Host Rock
<u>L1S13 - No. 1</u> 1.83 to 2.88 ft.	9,710 8,180	Disturbed Halite
<u>L1S13 - No. 3</u> 6.38 to 7.50 ft.	11,340 8,990	Anhydrite
<u>L1S13 - No. 4</u> 8.58 to 9.92 ft.	8,000 10,640 10,820	Undisturbed Halite

Table B-6. Data from Push-Out Testing on ESC Samples Removed from the PTM, 1991

Sample	Push-Out Strength (lbs/in² Shear)	Host Rock
<u>L1S13 - No. 1</u> 0.58 to 1.83 ft.	197	Disturbed Halite
<u>L1S13 - No. 4</u> 7.67 to 8.50 ft.	338	Undisturbed Halite
<u>L1S14 - No. 2</u> 2.25 to 2.88 ft.	117	Disturbed Halite
<u>L1S14 - No. 4</u> 6.50 to 7.42 ft.	267	Anhydrite Bed

APPENDIX C: PUSH-OUT RESISTANCE TEST METHOD

Scope. The push-out resistance test was devised to measure the force or pressure required to displace concrete or grout plugs cast in a rock medium. Measured values and visual inspection of fracture surfaces represent the quality of cement-based materials at the interface with the rock medium.

Significance and Use. Test methodology is similar to a modified shear bond test used to determine the bond and shear resistance between two concrete specimens produced by an epoxy binder. This push-out test can be performed on different shape specimens as long as the bonded surface of the plug remains parallel to the direction of applied force. Deviations of bonded surfaces from parallel would produce additional stresses not normally associated with bond-type tests. Conical or undulating surfaces increase the measured bond strength.

Apparatus. The test machine is a closed-loop hydraulically-actuated MTS materials testing machine. Load-controlled-type controllers applied a constant force or pressure to the plug instead of a displacement-type (screw-type) which applies a constant movement of compression platens with variable force. The interface between the concrete or grout and host rock is required to resist plug movement from the rock under constant pressure rather than constant movement.

The lower platen was a machined 1.0-in. thick hardened-steel compressive ring. Hardened-steel rings serve as reaction platens for the test specimen with the center portion removed to accommodate plug material movement. The inside diameter of the ring removed portion is machined to the plug diameter plus 0.25-in. allowing only the plug to be pushed out of the specimen. The outside diameter of the ring is machined to 2.0-in. plus the diameter of the test specimen.

Retaining sleeves were molded 0.0625-in. thick sheet metal cylindrical molds with clamping lips. Retaining sleeves serve to confine stresses allowing generated lateral stresses from vertical loading to dissipate. Allowing lateral stresses to dissipate into the mold prevents reaction against possibly weaker and brittle rock or the salt zone surrounding the specimen. Actual field plugs are surrounded or confined with the host rock or salt medium and the lateral forces acting upon plugs would distribute through the surrounding medium.

The upper platen was a machined piston used to push the plug out of the rock medium. The piston was the same diametrical dimension or slightly less than that of the plug. The piston lengths were required to be greater than 1-in. to accommodate the full breakage of the bond at the interface.

Test Specimens. Concrete and grout plugs were over-cored from the field site to include the host rock surrounding the plugs. The recovered cores removed from the site were extracted with coring devices having an inside diameter of twice the plug diameter (i.e. 2-in. diam. plugs were cored with a 4-in. diam. coring head). The test specimens were maintained in their as-received condition throughout the testing except during preparation for push-out testing.

Specimens selected for push-out testing were then sawed or ground smooth on each end. Sawing or grinding of ends were to within 0.001-in. planeness and normal to the plug to within $\pm 2^\circ$. Recovered core heights were maintained as much as possible. Final heights and diameters of both the plugs and host rock cores were recorded to the nearest 0.01-in.

Procedure. Place the test specimen on the ring platen. Center the plug in-line with the inside diameter of the ring. Slide the retaining sleeve over the test specimen and clamp in place. Caution must be observed to prevent movement of the specimen off alignment with the ring platen. Tighten the sleeve until it becomes snug around the specimen.

Center the test specimen with the upper platen of the test machine. Place the matching diameter piston directly in-line with the plug. Lower the upper platen to a contact position with the piston. Adjust the alignment as necessary between the ring platen, plug, piston, and upper platen.

Test the specimen in a load-controlled mode at the rate of 20 to 40 psi/sec based upon the diameter of the plug and not the entire core.

Calculations. The relative bond strength between the plug and host rock (S_b) is equal to the load (P_f) required to dislodge the plug from the host material divided by the contact area of the bond (A_c). The bond contact area is lateral surface area of the plug in contact with the host rock. For example:

$$A_c = 2\pi rl$$

r = mean radius of the plug as measured at the top and bottom of the plug

l = length of the plug as measured from the top to bottom

$$S_b = P_f \div A_c$$

S_b = calculated bond strength at the plug interface

P_f = observed maximum load to displace the plug from its host rock Results are reported to the nearest 10 lb/in².

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Office of Environmental Restoration
and Waste Management
Attn: M. Frei, EM-34 (Trevion II)
Washington, DC 20585-0002

US Department of Energy
Office of Environmental Restoration
and Waste Management
Attn: S. Schneider, EM-342
(Trevion II)
Washington, DC 20585-0002

US Department of Energy (3)
Office of Environment, Safety
and Health
Attn: C. Borgstrom, EH-25
R. Pelletier, EH-231
Washington, DC 20585

US Department of Energy (2)
Idaho Operations Office
Fuel Processing and Waste
Management Division
785 DOE Place
Idaho Falls, ID 83402

US Environmental Protection
Agency (2)
Radiation Programs (ANR-460)
Attn: M. Oge
Washington, DC 20460

US Geological Survey (2)
Water Resources Division
Attn: R. Livingston
4501 Indian School NE
Suite 200
Albuquerque, NM 87110

US Nuclear Regulatory Commission
Attn: H. Marson
Mail Stop 623SS
Washington, DC 20555

Boards

Defense Nuclear Facilities Safety
Board
Attn: D. Winters
625 Indiana Ave. NW, Suite 700
Washington, DC 20004

Nuclear Waste Technical Review
Board (2)
Attn: Chairman
S.J.S. Parry
1100 Wilson Blvd., Suite 910
Arlington, VA 22209-2297

Advisory Committee on Nuclear
Waste
Nuclear Regulatory Commission
Attn: R. Major
7920 Norfolk Ave.
Bethesda, MD 20814

State Agencies

Environmental Evaluation Group (3)
Attn: Library
7007 Wyoming NE
Suite F-2
Albuquerque, NM 87109

NM Bureau of Mines and Mineral
Resources
Socorro, NM 87801

NM Energy, Minerals, and Natural
Resources Department
Attn: Library
2040 S. Pacheco
Santa Fe, NM 87505

NM Environment Department (3)
Secretary of the Environment
Attn: J. Espinosa
1190 St. Francis Drive
Santa Fe, NM 87503-0968

NM Environment Department
WIPP Project Site
Attn: P. McCasland
PO Box 3090
Carlsbad, NM 88221

Laboratories/Corporations

Battelle Pacific Northwest
Laboratories
Attn: R.E. Westerman, MSIN P8-44
Battelle Blvd.
Richland, WA 99352

INTERA Inc.
Attn: J.F. Pickens
6850 Austin Center Blvd.
Suite 300
Austin, TX 78731

INTERA Inc.
Attn: W. Stensrud
PO Box 2123
Carlsbad, NM 88221

IT Corporation
Attn: R.F. McKinney
Regional Office
5301 Central NE, Suite 700
Albuquerque, NM 87108

Los Alamos National Laboratory
Attn: B. Erdal, CNC-11
PO Box 1663
Los Alamos, NM 87544

RE/SPEC, Inc.
Attn: W. Coons
4775 Indian School NE
Suite 300
Albuquerque, NM 87110-3927

RE/SPEC, Inc.
Attn: J.L. Ratigan
PO Box 725
Rapid City, SD 57709

Southwest Research Institute (2)
Center for Nuclear Waste
Regulatory Analysis
Attn: P.K. Nair
6220 Culebra Road
San Antonio, TX 78228-0510

SAIC
Attn: D.C. Royer
101 Convention Center Dr.
Las Vegas, NV 89109

SAIC
Attn: H.R. Pratt
10260 Campus Point Dr.
San Diego, CA 92121

SAIC (2)
Attn: M. Davis
J. Tollison
2109 Air Park Rd. SE
Albuquerque, NM 87106

Tech Reps Inc. (3)
Attn: J. Chapman
C. Crawford
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5000 Marble NE
Suite 222
Albuquerque, NM 87110

TRW Environmental Safety Systems
Attn: L. Wildman
2650 Park Tower Dr.
Suite 1300
Vienna, VA 22180-7306

Waterways Experiment Station (10)
Attn: L.D. Wakeley
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Vicksburg, MS 39180-6199

Westinghouse Electric Corporation (5)
Attn: Library
C. Conway
C. Cox
L. Fitch
R. Kehrman
PO Box 2078
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Westinghouse-Savannah River
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Universities

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Albuquerque, NM 87131

University of Washington
College of Ocean
and Fishery Sciences
Attn: G.R.Heath
583 Henderson Hall
Seattle, WA 98195

Libraries

Thomas Brannigan Library
Attn: D. Dresp
106 W. Hadley St.
Las Cruces, NM 88001

Government Publications Department
Zimmerman Library
University of New Mexico
Albuquerque, NM 87131

Hobbs Public Library
Attn: M. Lewis
509 N. Ship St.
Hobbs, NM 88248

New Mexico Junior College
Pannell Library
Attn: R. Hill
Lovington Highway
Hobbs, NM 88240

New Mexico State Library
Attn: N. McCallan
325 Don Gaspar
Santa Fe, NM 87503

New Mexico Tech
Martin Speere Memorial Library
Campus Street
Socorro, NM 87810

WIPP Public Reading Room
Carlsbad Public Library
Attn: Director
101 S. Halagueno St.
Carlsbad, NM 88220

Foreign Addresses

Studiecentrum Voor Kernenergie
Centre D'Energie Nucleaire
Attn: A. Bonne
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Atomic Energy of Canada, Ltd. (3)
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