

REPORT OF INVESTIGATIONS/1989

Backfill Properties of Total Tailings

By C. M. K. Boldt, P. C. McWilliams, and L. A. Atkins



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UNITED STATES DEPARTMENT OF THE INTERIOR Manuel J. Lujan, Jr., Secretary

BUREAU OF MINES T S Ary, Director

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UNIT	OF MEASURE ABBREVIAT	TIONS USED	IN THIS REPORT
in	inch	pct	percent
lb/ft ³	pound per cubic foot	psi	pound per square inch
m²/kg	square meter per kilogram	st/h	short ton per hour
min	minute	wt pct	weight percent
mm	millimeter		

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ABSTRACT

This U.S. Bureau of Mines report presents a study of three typical tailings samples as potential cemented backfill in underground mines. The testing series was unique in that the pulp densities of the samples were all above 75 pct solids. Test results included dry density; slump; percent settling after 28 days of curing; tensile strength after 28, 120, and 180 days of curing; and unconfined compressive strengths after 7, 28, 120, and 180 days of curing. The physical properties of the various test mixtures were further analyzed using linear and nonlinear statistical methods to produce correlations and mathematical equations. Physical properties were used to determine the influence of mix additives and as input for numerical modeling studies of backfill. The mathematical relations were used as a predictive tool in determining the suitability of various materials as backfill.

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INTRODUCTION

Conventional room-and-pillar mining has been commonly used in the United States. However, the domestic mining industry has been hardpressed to maximize mineral productivity in order to compete with foreign suppliers. As a consequence, U.S. mines no longer have the luxury of leaving ore-rich pillars as ground support, and reserves tied up in highly fractured material cannot be left behind. Existing mines are also encountering greater ground stresses as mining progresses deeper, causing the openings to squeeze inward dramatically or suddenly burst. In certain regions of the country and particularly near urban areas, ground subsidence poses safety and environmental hazards.

Backfilling stopes allows removal of pillars in addition to controlling ground subsidence. The fill also acts as a medium with established engineering properties and predictable behavior. These advantages can allow mines to maximize their ore reserves.

Backfill has been extensively used worldwide. The Bureau was involved as far back as 1964 in defining the properties of hydraulically placed backfill (1).⁴ From 1961 to 1970, Canadian researchers tested a multitude of mixes utilizing portland cement and mine tailings (2). These tests all used the common mode of hydraulic transport of materials that were typically <70 pct solids. At 45 pct water content, the backfill needed to be designed for high permeability where the excess water was drained and pumped out of the mine. Bleeding of the cement and aggregate fines through the drainage water was a common problem, resulting in greatly varied in-place strength.

Today, pumps and pneumatic blowers are capable of handling a mine's rugged environmental requirements while meeting a 100-st/h operating speed. These new pumps and pneumatic stowers may make it favorable to transport >80 pct solids, total tailings⁵ material from the mill to the stope for use as backfill. This capability no longer limits the mix matrix to 70 pct solids or to the inclusion of only the sands fraction of mill tailings. The resulting decrease in water improves the strength, homogeneity, and curing time of the material, and makes lean, cemented total tailings backfill an attractive option.

This report summarizes laboratory work done by the Bureau to define the strength characteristics of lean, cemented backfill using total tailings as aggregate, and varying the cement and other additives as well as the water content. The mix matrix used simulated the higher pulp densities capable of being transported and placed by large concrete pumps, physical stowing equipment, pneumatic blowers, or gravity free fall.

TEST PROCEDURE

MIX COMPOSITION

The mill total tailings used as the basic aggregate in this test series came from three underground metal mines: Tailings A from a deep silver mine in Idaho, tailings B from a lead-zinc mine in Missouri, and tailings C from a copper-silver mine in Montana. Grain-size gradation curves are shown in figure 1. The fines content of the total tailings was retained to minimize the void ratio. This practice has been documented as improving strengths and decreasing fill consolidation (3). Mix matrices are summarized in appendix A.

Commercially available Type I and II portland cement and tap water were used in all mixes in the test series. The following additives were incorporated in the test mixes to determine their influence on some of the physical properties of the tailings.

1. <u>FlyAsh</u>.-Various mixtures of commercially available ASTM Class F fly ash (4) were added to the tailings to determine whether the pozzolanic influence would be sufficient to decrease the required amount of cement and still maintain the unconfined compressive strength.

2. <u>Pit-Run and Ground Smelter Slag</u>.-The cementing influence of the smelter slag was determined by

Construction Technology Laboratories of Skokie, IL (5). The chemical analysis is shown in table 1. Since the hydraulicity, or the ability of the slag to react with water, is believed to increase when the slag is ground very fine (6), the tests included different gradations of ground slag. Slag samples of 400, 500, and $600 \text{ m}^2/\text{kg}$ as determined by the Blaine test (7) were mixed with water and showed no unconfined compressive strengths through 28 days of curing time because the material remained in the original slurry state. Grain-size analyses of the pit-run and ground smelter slag are shown in figure 2.

3. <u>Oil Shale Retorted Waste</u>.—Because previous oil shale research had documented the cementing properties of certain retorted wastes (8), oil shale retorted waste was used as an additive to determine if its cementing properties could be used in the backfill. The grain-size gradation curve of the retorted waste is given in figure 2.

4. <u>Kiln Dust</u>.-A locally available source of kiln dust was used in a few mixes to determine its cementitious effects.

5. <u>Superplasticizer</u>.-An ASTM Category B superplasticizer (9) was added to a few mixes to determine its ability to decrease the amount of water necessary to initiate cementing action and maintain pumpability. Ten times the

⁴Italic numbers in parentheses refer to items in the list of references preceding the appendixes at the end of this report.

⁵Total tailings, as used in this report, includes the full range of mine tailings, typically from 0.001 to 6.0 mm in diameter.







Figure 2.-Grain-size gradation curves for pit-run smelter slag, ground smelter slag, and oil shale retorted waste.

manufacturer's recommended dosage rate for concrete was needed before a measurable increase in slump was seen. Since results of the superplasticizer tests were not promising, no further tests were attempted.

Table 1.-Slag chemical analysis

<u>Major oxides</u>	Concentration, pct
Al ₂ O ₂	6.15
ΒắΟ [°]	.39
CaO	16.85
Cr ₂ O ₂	.05
CuŐ	.06
Fe ₂ O ₃	37.83
К,Õ ,	.68
MgO	2.09
MnO	1.81
Na ₂ O	.5
Ρ ₂ Õ ₅	.01
SíO ₂	30.95
TiO ₂	.37
ZnŐ ₂	.02

NOTE.-2.24 pct is accounted for by other concentrations or combustion losses.

MIX HANDLING

The mixing procedure for the test series included use of a portable cement mixer. After the oven-dried total tailings material and any additives were mixed for a minimum of 2 min and visually checked for homogeneity, a slump measurement was taken (10). Eight samples were taken from each test mix, packed into standard 3- by 6-in, waxed cardboard cylinders, and cured in a fog room (11). The slurry density was taken at the time of mixing, and the 28-day wet density was measured after 28 days of curing. Gang molds were cast using the various mixes for tailings B and C to obtain samples for determining tensile strength. These briquets were also cured in a fog room.

LABORATORY TESTING

The test series included unconfined compressive strength determinations after 7, 28, 120, and 180 days of

moist curing. Each strength test was run on a duplicate cylinder sample and the two strength readings were averaged to minimize errors. Eight tailings A cylinder samples were tested: two each for 7-, 28-, and 120-day cured, unconfined compressive strength tests (12); one for the 120-day cured, confined compressive strength tests; and one for determination of the dry density. Eight tailings B and C samples were tested: two each for 7-, 28-, 120-, and 180-day cured, unconfined compressive strength tests. In addition, the test results of three briquet specimens were averaged to determine 28- and 120-day cured tensile strengths for tailings B and C (13).

Initial mixes of the total mill tailings were cast without benefit of binder (cement, fly ash, etc.). These samples remained in a slurry state and did not achieve a compressive strength. In addition, the saturated environment of the fog room prevented any evaporation from taking place.

Appendix A summarizes the mix proportions along with the various additives, the types of tests conducted, and the test results. Cement, fly ash, pit-run smelter slag, ground smelter slag, kiln dust, and oil shale retorted waste were measured as a percentage of the total tailings aggregate (dry weight of fly ash plus pit-run smelter slag plus ground smelter slag plus kiln dust divided by dry weight of total tailings times 100). The water-to-cement ratio was calculated as a proportion of the weight of the water to the weight of the cement. The water-to-binder ratio was used to determine if the additives influenced the cementing properties of the mix and was calculated as the proportion of the weight of the water to the combined weight of the cement, fly ash, kiln dust, and oil shale retorted waste. Because the slag was known as a nonhydraulic additive, it was not included as a "binder."

The slurry density of the mixes was determined by dividing the weight of the solids by the weight of the solids and water. Slump measurements were taken to determine possible pumpability of the various mixes and was measured in inches. The tensile and unconfined compressive strengths (measured in pounds per square inch) were averaged through use of replicate testing.

TEST ANALYSIS

LABORATORY TESTS

Category B superplasticizer did not seem to have an impact on reducing the water content and increasing the workability of the mixes. This may have resulted from the nature of the superplasticizer used in the tests. The particular superplasticizer used was Mighty 150,⁶ a sulphonated

naphthalene formaldehyde condensate. This type of superplasticizer reacts by increasing the charge of the cement particle, thereby repelling the individual cement particles from each other and resulting in better dispersion throughout the mix. It also decreases the surface tension of the water, making it "wetter." In these lean mixes (<15 pct cement content), the influence of Category B superplasticizers would be diminished.

⁶Reference to specific products does not imply endorsement by the U.S. Bureau of Mines.

STATISTICAL ANALYSIS

At the beginning of the test series, tailings A results were statistically analyzed to determine if any meaningful relationships existed among the data. Ninety-five sample pairs were cast through the course of testing and included various additives such as pit-run smelter slag, ground smelter slag, and fly ash. Four candidate predicted variables (variables to be predicted from mixture information) were measured: average unconfined compressive strength at 7-, 28-, and 120-day curing increments and the resulting slump value. There were five predictor variables: pit-run smelter slag, ground smelter slag, fly ash, cement, and water-to-cement ratio. The predictor variables were mixed in varying proportions to test for fill properties of the tailings.

The Minitab statistical computer program was used for most of the analytic work and the primary statistical algorithm was a multivariate linear model (14). With 14 variables involved, it was necessary to perform a preanalysis of the variables that would sort out some of the more spurious prior to the multivariate model fitting. Therefore, pair-wise correlation coefficients between the variables involved were examined first. These values are summarized in table 2 and provide a quick method to determine which variables are most highly correlated.

The matrix in table 2 is a mixture of both predictor and predicted variables, as defined previously. An absolute value of 0.8 correlation coefficient was arbitrarily chosen to delineate significance (a 1.0 correlation coefficient is a perfect fit of the line to the data points). Using this criterion, 10 pair-wise relationships were deemed significant; however, three of these correlations were between the dependent variables themselves, i.e., 28-day average unconfined compressive strength (28DCOMP) versus 120-day average unconfined compressive strength (120DCOMP), with a correlation coefficient of 0.944.

Various combinations of variables were further analyzed by least squares fitting (15) a three-dimensional (3-D)hyperplane,⁷ which yielded the following equation:

$$7DCOMP = -73.6 + 32.4 CEMENT - 1.40 W/C$$

where

CEMENT = cement content, pct,

and W/C = water-to-cement ratio.

The analysis provided an R^2 value of 0.876.⁸ The various goodness-of-fit parameters r, R^2 , and I are discussed in

appendix B. The results of the linear regression analysis are listed in appendix C.

Extreme scatter in some of these data is apparent in the itemized predicted variable (predicted Y-value, Fit column) versus the actual data (7DCOMP column) in appendix C. To illustrate, observation 30 of appendix C lists an actual 7-day observed strength of 160 psi. However, the predicted value using the regression equation produced a result of 343.35 psi. For this reason, two other modeling schemes were investigated: a multivariate, linear stepwise regression model and a univariate, nonlinear exponential model. The differences between these models are described in reference 15.

To determine a best multivariate linear model, stepwise regression was applied to the tailings A data. Briefly, this is a procedure that picks the predictor variables one at a time in order of relative importance. This approach has two advantages to the user: it produces a linear model to represent the data, and in so doing, it searches for the most important subset of dependent variables that will do the job. The Bureau's stepwise code has an additional advantage in that it allows the creation of variables that are derived from the original predictor variable set. For example, cement and fly ash content were predictor variables. Terms involving cement or fly ash squared, cubed, multiplied, raised to powers, etc., can be easily inserted in the model. There is one important aspect, however, which must be kept in mind when using this model. In forming the regression, the user is always fitting an additive model of the terms of interest.

The stepwise procedure was applied individually to each of the predicted variables involved: 7-, 28-, and 120-day cured, unconfined compressive strengths and the slump variable.

Mathematical representation of the stepwise regression model is given by

$$Y = P_1 + P_2 Z_1 + P_3 Z_2 + P_4 Z_3 + \ldots + P_1 Z_k,$$

where

$$P_1, P_2, \dots, P_i$$
 = constants found by the stepwise process,

and

$$Z_1, Z_2, \ldots, Z_k$$
 = selected predictor variables (cement,
fly ash, etc.) chosen one at a time
in order of importance.

It was necessary to use four predictor variables (cement, pit-run smelter slag, water-to-cement, and fly ash), of which only cement and pit-run smelter slag were deemed statistically significant, to produce an equation predicting the 7-day unconfined compressive strength with an R² value of 0.887 (table 3).

⁷A three-dimensional linear model.

 $^{{}^{8}}$ In all multidimensional analyses, R² refers to the multivariate correlation coefficient (appendix B).

Table 2.-Linear correlation coefficients for tailings A variables

		CEMENT	FLYASH	PRSLAG	GRSLAG	W/C	W/B	SLUMP	SLDEN	DRYDEN	WETDEN	SETTL	7DCOMP	28DCOMP
FLYASH .		0.106							•	· · · · · · · · · · · · · · · · · · ·		-		
PRSLAG .		.260	-0.222											
GRSLAG		.353	170	-0.222										
W/C		873	059	198	-0.259									
W/B		705	572	060	142	0.758								
SLUMP .		336	.232	314	540	.283	0.057							
SLDEN		.487	099	.561	.477	388	257	-0.870						
DRYDEN		.398	102	.473	.457	548	408	726	0.820					
WETDEN		.315	519	.557	.353	325	015	603	.653	0.701				
SETTL		274	201	113	081	.143	.253	.334	328	096	0.001			
7DCOMP		.939	.168	.165	.325	821	680	346	.456	.432	.288	-0.255		
28DCOMP		.921	.182	.120	.389	781	671	358	.456	.450	.283	254	0.977	
120DCOM	Ρ	.848	.326	.023	.409	738	697	330	.420	.443	<u>.</u> 166	288	.930	0.944
CEMENT	Cement conte	ent, percent o	of tailings.		SLDEN	Slurry o	density.							
FLYASH	Fly ash conte	nt, percent of	f tailings.		DRYDEN	Dry de	nsity.							
PRSLAG	Pit-run smelte	er slag conten	nt, percent of	tailings.	WETDEN	Wet de	nsity.							
GRSLAG	Ground smelt	er slag conte	nt, percent o	of tailings.	SETTL	Percen	t of settling].						
W/C	Water-to-cem	ent ratio			7DCOMP	7-day c	compressiv	e strength.						
W/B	Water-to-(cerr	nent-plus-fly a	ash) ratio.		28DCOMP	28-day	compressi	ve strength.						
SLUMP	Inches of slur	np.			120DCOMP	120-da	y compres	sive strength.						

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Model	Mathematical equation	Goodness- of-fit measure
3-D hyperplane	7DAVG = -73.6 + 32.4 CEMENT - 1.40 W/C	$R^2 = 0.876.$
Multivariate, linear stepwise regression	7DAVG = -82.58 + 33.6 CEMENT - 0.74 PRSLAG	$R^2 = 0.887.$
Nonlinear, exponential	7DAVG = 1,515.53e ^{-0.57} W/C	I = 0.889.
7DAVG 7-day cured, unconfined compressive	strength. B ² Multivariate correlation coefficient.	

CEMENT Percent cement in total aggregate contained in mix. W/C Water-to-cement ratio.

Another modeling attempt was made using a twodimensional model with the predicted and predictor variables fitted by an exponential curve. Figure 3 illustrates the data by plotting the 7-day unconfined compressive strengths to the water-to-cement ratio. This curve-fit procedure resulted in the following equation:

$$7DCOMP = 1.515.53e^{-0.57} W/C$$

where

7DCOMP = 7-day unconfined compressive strength, psi,

W/C = water-to-cement ratio.



Figure 3.-Seven-day compressive strength versus water-to-cement ratio for tailings A.

PRSLAG Percent pit-run smelter slag contained in mix.

Index of determination.

Table 3 tabulates the various results of the three statistical methods used to determine goodness-of-fit for the tailings A test data as applied to predicting the 7-day unconfined compressive strengths. The 3-D hyperplane produces a correlation coefficient of 0.876. In the multivariate, linear stepwise regression model, the anticipated 7-day unconfined compressive strengths fit the observed compressive strengths of each test specimen with a correlation coefficient of 0.887. The predictor variables, listed by order of importance to determine the 7-day compressive strength, are cement and pit-run smelter slag. The nonlinear, exponential model produces an index of determination (see appendix B for definition) of 0.889, which is quite promising since it is based on only one input variable, the water-to-cement ratio.

After it was determined that the exponential model would best fit the data curves of the unconfined compressive strengths versus the water-to-cement ratio, the data from tailings A, B, and C were analyzed as a group for comparison. Plots of the compressive strengths versus water-to-cement ratios for the total data base are presented in figure 4. The mathematical representation of the curves along with their respective indices of determination are given in appendix D.

Further analysis of the tailings A, B, and C data included exponential curve fitting of the compressive strengths to water-to-cement ratios for the mixes grouped by tailings type and then by tailings type not containing any additives (pit-run smelter slag, ground smelter slag, fly ash, kiln dust, and oil shale retorted waste) (figs. 5-6). The mathematical representation of the curves along with their respective indices of determination are also given in appendix D.

As can be seen in appendix D, the goodness-of-fit increases as the data base becomes increasingly selective. For instance, the total data base index of determination, I, for 7-day compressive strength is 0.796. For the data base containing only tailings A, I is 0.889; and for the tailings A data base not containing any additives (pit-run smelter slag, fly ash, etc.), I is 0.982.



Figure 4.-Compressive strengths versus water-to-cement ratio for total tailings data base.



Figure 5.-Compressive strengths versus water-to-cement ratio for tailings A, B, and C.



Figure 6.-Compressive strengths versus water-to-cement ratio for tailings A, B, and C (not containing additives).

DISCUSSION OF RESULTS

The results indicated that the addition of oil shale retorted waste without the benefit of cement produced compressive strengths on the order of 100 psi in 28 days. The cementing properties of the retorted waste were greater for the finer particles of tailings C. The addition of fly ash improved the compressive strength of the total tailings aggregate. As the tailings grain size fraction greater than 200 sieve increased, the influence of the fly ash decreased. The 28-day compressive strength of tailings A was increased by 25 pct, tailings B by 48 pct, and tailings C by 98 pct over the compressive strengths gained by the use of cement alone.

As the grain size of the tailings fraction greater than 200 sieve decreased, the compressive strengths also decreased for the various curing periods. The 7-day compressive strength for tailings A with 6 pct cement and a water-to-cement ratio of 4.5:1 was 118 psi; for tailings B it was 107 psi; and for tailings C it was 65 psi.

The linear relationship (based on least squares fitting) between the 7-day compressive strengths and those of 28, 120-, and 180-day compressive strengths is presented in figure 7. In each case, the strength gained between each pair of relationships was greater as the grain size of the tailings material increased. This was just the opposite for the relationship between the 7-day compressive strength and the 28-day tensile strength (fig. 7). There was no significant difference in strength gain between the 28-, 120-, and 180-day compressive strengths and the 7-day compressive strength because of grain-size differences (fig. 8). However, grain size differences caused a marked difference between the 28-day tensile strength and the 7-day compressive strength (fig. 8). The finer-grained tailings C developed a higher tensile strength when compared to the 7-day compressive strength.

The ratios between compressive strength to tensile strength for the various days of curing ranged from 4.4 for



Figure 7.-Seven-day compressive strength versus 28-, 120-, 180-day compressive strengths, and 28-day tensile strength for tailings A, B, and C.

the total tailings not containing additives to 4.8 for the total data base.

The goodness-of-fit for calculating the compressive strengths using the water-to-cement ratio and the exponential formula is

$$Y = Ae^{-BX}$$
,

where Y =compressive strength, psi,

X = water-to-cement ratio,

and A and B are constants.

This goodness-of-fit progressively improves as the sample groupings become more restricted. Appendix D summarizes the indices of determination for the various groupings.



Figure 8.-Seven-day compressive strength versus 28-, 120-, 180-day compressive strengths, and 28-day tensile strength for tailings A, B, and C (not containing additives).

SUMMARY

This series of backfill material testing was initiated to determine what engineering properties could be expected from a variety of mill total tailings. The incorporation of additives was meant to define the extent of increased strength or workability of the resultant mix. None of the tailings tended to be self-cementing. However, as cement contents were increased, compressive strengths increased. The compressive strengths of the fly ash-and-cement combination increased after 28 days of curing as compared to the strength of cement alone after 28 days. The addition of pit-run smelter slag, which incorporated coarser particles into the mix, seemed to increase compressive strength, but the slag alone was not cementitious. In some cases, such as those where oil shale retorted waste was added to the tailings, a full range of mixes was not attempted since the problem was merely to determine whether or not the retorted waste was a detriment to the mix, thereby indicating possible uses of oil shale waste.

Further research will test the relationships found during this investigative test series. The effects of chemical additives such as superplasticizers, high-early-strength cements, water-reducing agents, and kiln dust will be examined. With further refinement, an accurate predictive tool will be developed that will assist the industry in analyzing the suitability and stability of dewatered, total-tailings backfill. 1. Nicholson, D. E., and W. R. Wayment. Properties of Hydraulic Backfills and Preliminary Vibratory Compaction Tests. BuMines RI 6477, 1964, 31 pp.

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APPENDIX A.-MIX MATRIX AND TEST RESULTS

Sample		Addıtiye	wt/taili	ngs wt, pc	t		W/B	¥/C	Slurry	Dry	28-day	Slump,	28-day	Av tensil	e strengt	h, psi	av uncost	lines compre	ssive streng	th, ps1
80.	Cament	Fly	Pit-run	Ground	Kiln	Retorted			density,	censity,	wet	រោ	Settlement.	28 day	120 Cay	180 day	7 day	28 day	120 day	180 day
		ash	slag	slag	dust	waste			wt pct	lb/ft3	density,		pct	}						
											16/ft3									[
	L			1		L					7471 1107		L	L	<u> </u>	<u>}</u>			L	<u> </u>
			0	۵	0				7/	50	HILLINDS	8		No		80	10			
	4	v		. v	יי מ	V	0.J 6 /	a E	, ;o 76	70	123	11	2.1	19H 100	64 14	an No	70	12	27	NR
2	6	U A			0		3.0	2.0	13 1 73	100	125	11	5.4	: NH 1 NA	RH NA	NH NA	70	110	121	NA
2	9	6		· v	ų	e e	4.4	3.4	. /6	191	129	11	0.0	NH NH	7 8	MH HA	112	157	205	NH.
4	10	0) U		9	0	3.33	2.33) // 	102	123	11	J	: NA	NK	NH	212	318	411	NA
3	12	C.	; i	, ,	0	ų į	2.78	2./3	1 1/	100	124	11	4.3	· NS	98	RA MA	278	405	4/5	NS
6	14	Ű			ų	· U	2.08	2.30	1 <i>11</i>	102	126	11		o N∩ ∙	N R	NH	J40	597	385	NA
1	4			/ V			5.3	5.1) 63 - 61	107	124	1.1	· · · ·	r NH	719	NA	/4	44	58	NA
8	4	· · ·) V N A		/ 12 17 0	0 4 5		2 51 2 51	59 1 100	120	2.7	2	L 1959 D NA	76 86	NA NA	48	<u>b</u> 1	91	NA
7	· •		4 1 1	, v , a) U	0,5	0,. 7	. QV 5 07	100	171	•••		с ли 5 ил	14H 21A	RH NA	34	35	101	NH:
10	· 0		х (с)	/ U		, v , a	ل بر ۸	94.	J DJ 4 D-1	110	101	***		: ,11H I NA	: (TH NA	NH NA	170	277		24
11	. 3	· ·	۰ ۱) (\ a	4 A		- 01 - 01	119	· 120 · (57		- 2	אות ד או כ	ч <u>л</u> на	84 81 A	171	180	280	8H No
14				v v a r		9 9 5 6	5 · 2	7.	J 00 7 70	10/	12/	5.;	.د. د ۱	ດ ກປະ ດ ນ/		NR MA	118	170	206	NH
1.			i .	0 i			3 0/	1.7	2 /C E 70	100	1 127			2 N/ E 117	1 AN	1929	23	131	404 705	NA
1.		i 1		0 i		0 V 0 0	1.00	1.7	J /C 1 70		(143 (175	. A.		3 NF 4 M/	/ NH	NH MA	156	170	572	NA
1.		, 1	1	0 U	,	v v A A	1.00	1.1	4 /0 7 70		1 124		1	אח ד עו ד	е пл м	лн ли	171	100	010	NH
14	> 11 7 • 1		1	v (0 (، ۶ ۱	0 0 0 8	1.77	2.1	לז ; סד מ	• • • • • • • • •	t 104		1 7	ು ಮೇ ೯ ಟು	। म्स . स्ट	NA NA	471	952	9/4	NH
1.	1	1 C	4 4	0 1 0 1	/ 1	0 0 0 0	1.00	2.7	ຜ :; 5 70	·	2 124	· 7.	2 C		t 1414 N 1413	84 84	431	701	1,241	N8 NA
1		e 1	1 E	0 (6 /	· ·	0 0 0 3	1,17	2.1	4 /: 6 0/	101	2 121 7 175	· 7.	- -	ও ন। ত ম	т пн \ жл	NP(140	101	513	1,2%3	NH
1	7 C	5 ∠ 7 7	.J .F	ν ι Δ	<i>,</i>	0 0 0 0	1+1	D,	0 01 1 0/	10	1 12. T 177	10.	J 4.	4 NS 7 N	1 112A N NA	: ЛН 	114	213	359	AH
<u> </u>		5 <u>4</u>		9 4 9 4	, n	9 V 0 0	1,92	۹. 7	ୟ ପ୍ୟ ମ ପ୍ୟ	7 10.	(124 6 17-		v 7 1	ວ Ni 7 ນ	1 AM 6 NA	. NH	135	200	401	A H
4	1 1	v 2		0 0		0 0 A A	0.79		v 0	2 LL 1 - 11	2 10 1 1 1	· 7.	/ 7.	ง พ ร.น.	1 NH	1 NH	207	453	840	
4	21 7.	2 <u>4</u> e 7	ີປ	0 1	4 5	0 U	0.00	4.0	יד ס. ה ה	t 44 1 1 1	7 12. 7 1.0	-	o /	4 % 5 N	1 Ar 	1 NH	407	127	1,380	· #18
4) 1 • •	24	(3)) E	0	v 0	0 U 0 A	0.00	24 1 0		1 11	3 12. 8 47.	, ,	0 5	с N л н	4 NA 6 NA	4 DHA	219	1/6	1,744	NA NA
2	4 1 F	e 1	10 A	ų n	U =	0 0 0 A	V.20	1,0	17 G 6A 7	1 34 1 34	7 123 A 14) /.		2 N	H NA 7 44	1 NH 3 NA	695	1,024	4	I NH
4	2	9 4	v	0 ·	J E	γ γ Λ Λ	0,_/		17 7. 17 7	0 // 7 10	H พ. (เว	1 <u>1</u>	.L P	и н г н	N 01	1 NH	14	1.		NA NA
4	0 7	•	0	۰ ۵		v v N 0	:,20	/.: E	10 7 10 7	/ 10 7 to	1 12	1 1 7 7	1 1	1 A	את א וע א	1 NH 1 HA	47	10	115) NH
4	/	0 n	0	v č	J 6	V V 0 1	. 7.07	J 7	24 7 70 7	7 10 D 16	7 14 8 10	1 I I		2 N	ы п. А н	n 16M NA NA	7J	1.00	للذ / درج ج	. 10H
4	5 n 1	a (v 0	v 0	5 5	0 0	/ ⊋•7∠ / ₹13		12 /	0 10 D 10	w 12 1 17	ני ס ז ו ר	:1 J. 1	2 M	н п • •	n, AM 1 M/A	130	1 42	· 44:	14N C
	7 1	.4 .7	0	v A	5	a a	211		10 ; 11 7	a 10 a 10	0 12	/ 1 9 1	11 T	и и и и и	n n. A N.	н лен 6 NA	100	, JJ.) 36') 36'	г 144 Э Ма
	1 V I	3 3	Q 0	0	e e	v .	2.01	7	14 7 74 7	a ii	0 12 0 13	1 1	1 5		ता स २. स	n an c Mò	100	5 <u>5</u> 6	J JD JD	1440 S
	u 1 19	т 0	0	0 1	1	0 (0 f	- 21-30 NG		50 7 NG 7	7 17	5 IJ	3 . 2 :	11 4	.0 m t k	n n A N	n na N NG		2 Jei) i	2 /D- \ A1	ייים ג אוג 1
	12	٠ ٨	0	0 1	4	0 1	, jun 1. 7.113	7	αο / Δ' 1	7 12 9 17		a i	11 5	7 1	н н 10 М	n an A NG	. 77	י י		/ 1949 6 NA
	10 Ca	7	0	0 I	.1	й (1 4 75	4	95 7	9 1/	12 17	6 	11 4	··· ·	เก ส เอิ ม	n an A bù	, /2 	2 U 7 (7	2 10	5 NA 7 NA
	(f)	, 0	¥ ň	v 1 6 1	1	0 1	v 7.71	, 1. T	71 7	5 . Na 10	11 21	4	11 K	7 1	10 V	а ал А NA	1.25	5 12	J 31. 7 87.	2 100 D NA
	1. 1. 1	, ,	n n	0 1	it.	0 1	, 997 797		97 5	10 II	10 17	e A	Q 4	4 1	67 N	n ⊪r ∆ N2	10	J 145	, 10 7 49	2 146 5 80
		17	τî 1	ò i	11	ŭ .	5 2.49	2.	48	19 ti	10 12	7	9 4	.4	10 N	A 11	1 374	a 57	u A0	5 NS
	το .	16	ů.	0 1	11	Å i	n 215	2.	12 5	10 1	17 15	7	9 4		10 N	a 122	100	, ., a .,	40 UU A 97	u (16) 5 M2
	70 . 70	5	ñ	0	33	0	о <u>5.</u> 25		25	 	12 1	4		4	VG h	10 N	. , <u>,</u>	9 15	a 32	5 N
	40	B	0	0 :	33	0	0 4.17	· 4.	17	1 1	14 13	1. 16 5	.5	4	NA N	IA NI	A 13	1 76	3 43	4 N/
	41	11	0	0	33	0	0 3.13	5 3.	.13	BI 1	18 1	29 4	.5	2	NA N	10 W	A 29	- 49 5 52	3 1.09	
	42	13	0	0	33	0	0 2.	5 2	.5	81 1	15 U	57 5	.5 3	.5	NA N	IA N	4 34	1 44	4 1.07	- N
	43	16	0	Ð	33	0	0 2.0		.08	92 1	16 1	39	3 7	.7	NA N	(A N	4 57	- Q3	a 1.39	- 11 9 11
	44	19	0	0	33	0	0 1.7	9 1.	79	B2 1	18 1	39 7	.2	2	NA I	NA N	A 77	2 1.39	2.6	G N
	45	8	0	0 1	00	0	0 4.1	7 4	.17	86 1	22 1	41	0 3	2.5	NA 1	1A N	A 12	5 75	1 5	5 N
	46	12	0	0 1	00	0	0 2.7	8 2	.76	85 1	21 1	28	0	2	NA I	Ka N	A 27	10 3 <i>1</i>	13 B4	12 N
	47	té	0	0 1	60	0	0 2.0	8 2	.08	97 1	27 1	32	0	1	NA	NA N	A 45	ii 9:	2.10	- 6 N

See explanatory notes at and of table.

MIX MATRIX AND TEST RESULTS-Continued

Sample		Additive	wt/tailis	45 #t. pc	:		1172 1172	910	Same	fie :	70	<u></u>	<u> </u>		_		r			
π۵.	Cement	Fly	Pit-run	Sround	Kiln	Retortes	1		den citv	densa	25-03V	, 2187b°	2c-day	Av tensil	e strengt	h, psi	Av ພກເອກ	fined compre	ssive streng	th, p≤1 -
		asn	siag	siag	dust	waste			at ocr	12/443	date: ty	10	sectiement,	18 GAA	120 cay	180 day	⁷ day	28 day	:20 day	180 Cay
											16/6+3		μ							
	L	L	_	L	ł						10/100			L	Ĺ					
								-			TAILIN65	A						·		
40 80	29	0	ų A	100	ġ.	0	1.07	1.67	87	126	107	0	4.5	NA	NA	N0	407			
47	++ -0	0	0	100	0	Û	1.39	1.39	37	125	130	0		NA	NA	NG.	97. 74.5	637	1620	hA
51	20	0	<i>'</i> j	100	0	0	1.19	1,19	87	126	139	0	3	NA	AR	NA.	777 202	1,002	1,740	NA
51	U 4		0	13	0	0	NA	NA	77	122	NA	11	4.1	NA	NA	NA	000 11	1,/*/	2,144	NA
57	5	13	U A	13	U .	0	1.89	6.0	80	109	123	11	3.3	NA	NA	NA	39	174	0	ХH
51	10	13	ů v	10	U A	U A	1.65	4.4	50	109	114	11	2.5	NA	NA	NA	154	230	200	4 8
55	17	13	0 0	1.5	0	0	1.47	2.3	80	112	120	9	2.8	hŔ	NA	NA	264	507		NR US
55	15	13	0	13	0	0	1.52	2.54	81	113	125	9	2.3	NA	NA	NB	341	770	1 97	/4H •/*
57	18	13	Ő	13	0	0	1.2	2.2	81	113	122	8.7	3.5	NA	NA	NA	501		1.454 7.454	л. ИЛ
58	5	13	13	10	v 0	0	1.1	1.89	81	115	124	8.2	0.7	NA	NA	NA	357	639	947	3 4 *2
59	8	13	13	0	0	ů ů	1.07	0.0	80	109	123	11	2	NA	NA	NA	152	281	578	NA
50	10	13	13	Ô	ů v	ů.	1.51	4.4	80	107	123	11	2	NA	NA	NA	89	162	237	XA
61	13	13	13	0	ň	ő	1.77	2.3	00	110	122	10.5	2.3	NA	NA	NA	233	45a	851	NA
62	15	13	13	0 0	0	0	1 2	2.04	01	111	125	10.5	1.3	NA	нA	NA	363	533	1,165	NA
63	18	13	13	9	n D	0	1.1	1 89	01	114	122	10	2	NA	NA	NA	492	782	1,449	NA
64	9	0	11	0	Q	0	44	NG.	77	113	120	7.3	1.2	NA	NA	NA	623	991	1,988	ни
65	4	Û	11	0	Ô	0	7.42	7.43	79	104	195	11	. NA	**	NA	NA	0	0	. 9	NÁ
66	7	0	11	0	Û	0	4.95	4,95	79	105	125	11		NA	NA	NA	71	93	121	NA
ь?	Ģ	ð	11	0	0	0	3.71	3.71	78	105	120	11		89	NA	NA	134	195	263	NA
66	11	Û	11	0	0	0	2.97	2.97	79	103	179	11	1./	8A	NA	NA	179	224	410	NA
69	13	0	11	0	0	0	2.48	2.48	79	105	NA	11	2.7	NH NA	NA	NA	307	377	600	NA
70	16	Û	11	0	0	0	2.12	2.12	79	106	NA.	11	J.2 1 7	NH NA	NA	NA	292	380	594	NA
74	4	0	11	0	0	0	5	5	84	118	134		1.7	RH NA	MH NA	NA	451	633	866	NA
72	4	0	11	0	0	0	5.5	5,5	83	116	129	<u>م</u>	2.1	MH MA	RH HA	NA	130	173	225	NA
13	4	0	11	0	0	0	6	6	81	111	128	7.5	2.1	NA NA	AH NA	54 NA	113	130	180	NA
/4	1	0	11	0	0	0	3.5	3.5	83	119	133	2.2		NA NA	ня 4А	NH	88	115	147	NA
15	/	0	11	0	0	0	4	4	B2	114	132	5	4.2	NÓ	NA NA	nn NA	215	309	377	NA
10	/	U	11	0	0	0	4.5	4.5	80	NA	131	10.5	2.3	NA NA	NA NA	MA NA	137	208	266	NA
71	U E	U	50	0	0	0	NA	NA	80	127	133	NA	16.8	NA	NA	NΔ	103	155	281	NA
70	J	0	زن 77	U	0	0	6.25	6.25	81	111	133	11	4.5	NA	NA	NA	0 00	150		NA
80		0	נג זז	v	0	0	4.17	4.17	81	113	135	11	4.3	NA	NA	NA	140	210	247	NA
81	13	0	33	Ň	0	U	3.09	3.09	81	117	136	9	3.5	NA	NA	NA	441	210	000 000	NH
82	15	Ň	55 77	0	0	0	2.48	2.48	82	115	134	10	4	NA	NA	NA	325	490	60V 747	NA
83	19	ŏ	33	ň	0	0	2.06	2.06	82	115	135	9	3.2	. NA	NA	NA	466	674	914	ан Мо
84	4	0	33	ň	0	0	1.17	1.77	82	117	136	8.7	2.7	NA	NA	NA	668	966	1 789	80 10
85	4	0	33	ŏ	ő	0	5 5		84	121	NA	3.5	4.2	NA	NA	NA	107	138	724	NA NA
86	4	0	33	0	0	0	J.J 4	J.J 1	183 01	NA 110	NA	5.7	NA	NA	NA	NA	99	147	220	NA NA
87	В	0	33	Ō	Ő	ŏ	3.5	रन	01 97	112	716. NA	8.5	3.4	NA	NA	NA	69	116	159	NA
88	8	0	33	0	Ó	0	4		60 60	110	NA	2.2	3	NA	NA	NA	149	238	371	NA
89	8	0	33	0	0	0	4.5	4.5	92 R0	113	11H NA		5.3	NA	NA	NA	122	168	278	NA
90	8	0	100	0	0	Ō	4.13	4.13	84	177	140	11	NA	NA NA	NA	NA	80	153	195	NA
91	12	0	100	0	0	0	2.75	2.75	87	125	140	2.0	1.8	NA	NA	NA	125	179	262	NA
92	16	0	100	0	0	0	2.06	2.06	87	176	137	1 7	1.8	NA	NA	NA	216	357	481	NA
93	20	٥	100	0	Ð	0	1.65	1.65	87	125	141	1.7	2	. NA	NA	NA	322	528	698	NA
94	24	0	100	0	0	0	1.38	1.38	87	123	171	2.3		NA	NA	NA	420	719	1,017	NA
75	28	0	100	0	0	0	1.18	1.18	87	123	140	2.7	1.5	NA NA	SA MA	NA	610	938	1,342	NA
_ .												2.0	2.0	nHr	MĤ	NA	824	1,300	1,799	NA

۰ .

See explanatory notes at end of table.

MIX MATRIX AND	TEST RESUL	TS-Continued
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Sample		Additive		ngs wt. pa	:t		¥/B	870	Slurry	êry	28-dav	Sluep,	28-day	Av tensi	e strengt	h, psi	Av uncont	fined compre	ssive streng	th, psi
no.	Cement	Fly	Pit-run	Ground	Kiln	Retorted	1		density,	density,	wet	LſI	settlement.	28 day	120 day	180 day	7 dav	28 jay	120 aay	180 day
		ash	slag	slag	dust	Raste	1 1		wt act	16/ (t)	density,		pet	}	}					
		1									lb/ft3									
<u>4</u>				1	al		-L		L	L	TATI INCO	۰		L		1				
	1		0 0	0		0 0	55		5 87	NA	17.1	<u>~</u>	1 1	2 10	75	۵Ľ	29	126	151	μA
2			ω 0 0 0	, v		e v		3.0	/ 00 . ai	500 500	102	0.75	- 712 - 712	L 10 2 20	19	84 MA	72 40	120	1	лн 85.
÷	4		0 0 0 0	. a		n n	د د		5 9A	No	129	9 4	5 54		19	NA NA	5	50	31	NA NA
ر ه	۲ ۵		v v 6 1	, , , , , , , , , , , , , , , , , , ,		γ γ h A	3 5	, i	5 87. E 87.	NA NA	136	2.7	ः २,२ २ २,१	2 19 2 19	5.9	NC.	190	275	11	AH No
			ε 0	0		ů O	4		1 82	NA	128	4.5	5 4.3	2 74	39	NA	154	204	795	54
-			ν	, î		n n	4.5	4.5	5 BÓ	NA	120	1	0 4.9	34	40	44	107	144	(21	44
7	3		, a (, -) 9		0 0	2.5	2.5	5 84	10	131	1.	5 4.5	5 89		NA	356	456	671	NĂ
9	6	, }	ù () 0		0 0	5		32 32	NA	127	4.3	5 3.1	6 64		NA	270	255	441	NA
-	8		0 () ຍ		n 0	1.5	3.	 קיי 5	56	(27	1	0 5.3	2 52	66	VA	175	201	763	NA
10	10		0 () ຢ	,	J D	2		2 85	NA	131		1 4.	5 113	153	88	551	764	923	YA
11	10)	υ (, ()	0 0	2.5	2.	5 81	NA	128	5.	5 4.5	5 89	107	44	365	526	692	NA.
15	10)	υ i) ()	0 9	3		3 19	NA	125	1	1 4.	5 56	87	NA .	225	340	421	NA
1]	12	2	0 (0 1)	6 0	1.5	1.	5 84	NA	1 133	1.7	53.	1 133	171	NA	716	952	1.231	NA
14	12	2	e ()	0 0	2		2 82	N#	134	4.2	5 3.1	8 12	2 160	NA	535	733	937	NA
1;	5 13	2	0	0 (9	0 0	2.5	2.	5 79	N/	134	1	1 5.	3 8	3 116	NA NA	325	440	528	NA
14	5 14	4	0	0 ()	0 0	1.5	1.	5 84	N/	137	1.	5 4.	2 18	244	NA	955	1,482	1,534	NA
1	4.5	5 1,	5	0 (0	0 0	3,5	4.6	7 83	i Ni	136	2.	5 4.	52	7 55	5 47	116	193	275	313
1	a 4.1	51.	5	Ų I	0	0 0	4	5.3	3 83	2 Ni	A 134	1 7.	5 4.	5 2	5 47	41	96	141	295	191
14	9 4.	51.	.5	0 (0	0 0	4.5		6 8) NE	4 13)	. 1	0 6.	2 1	7 39	5 44	81	114	214	194
2	<u>ن</u> ک	ь	2	9	D	0 0	2.5	3,3	13 B	i N	A 132	1-2	25 4,	2 5	5 112	2 105	110	200	358	205
2	1 1	6	2	0	0	0 O	5		4 5:	2 Ni	4 131	l.	6 5.	.6 N	é 64	4 91	130	223	404	381
2	2	6	2	Û	0	0 0	3.5	4.8	יז ק	? N	A 134	1 1	1 7.	С Н	A 5;	1 42	194	175	152	297
2	37.	5 2.	.5	0	0	0 0	2	2.8	7 8	5 N	A 13	2	2 4.	.2 N	A 14	9 135	278	371	786	830
2	4 7.	52	.5	Q	0	0 0	2.5	3.3	3 8	1 N	A 12	1	8 3.	.I N	A 74	9 55	168	162	577	611
2	5 7.	5 2	.5	0	0	0 0	3		4 7	7 1	A 13	, ,	11 7.	.3 N	A 6'	2 66	134	212	333	2 371
2	6	9	3	Ð	0	0 0	2.5	3	33 7	9 N	A 12'	9	11 7.	.3 N	A 10	0 111	190	314	519) 5 <u>0</u> 9
2	7	9	3	0	0	0 0	2	2.0	a7 8	2 N	A 13.	2 4.3	75 5.	.2 8	8 14	1 176	235	417	671	765
2	8	3	3	0	0	0 0	4		8 8	2 N	A 13.	2	9 7.	.3 1	3 2	7 30	50	99	193	2 211
2	4	2	2	0	0	0 0	5		12 8	1 N	8 13	1 8.	.5 7.	.3	0	0 0	36	57	7	5 75
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	43	-~ 6	6	0	0	0	0 6.2	2 12	47	50 1	KA 12	20	11	NA	22 1	22 1		7/ A 7	יים ד 1	., JA 13 10
	44	6	6	0	0	0	0	4	8	70	NA 12	26	11	6	31 0	71 50	Na	A 10	7 25	2 783
	45	6	6	0	0	0	0 2.3	3 4.	ь 7	80	NA E	56	11 7	.3	50 I	35 101	N N	A 20	2 4	5 52

See explanatory notes at end of table.

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MIX MATRIX AND TEST RESULTS-Continued

iample		Add	tive w	t/taili	ngs wt,	pct			/8	₩/C	Slurry	Drv	28-day	Slump,	28-cay	Av tensi	le strengt	h, p51	Ay ancon-	ined compre	ssive streng	th, ps1
n o.	Cenent		Fly asn	Pit-run släg	Sround slag	K1 du	ln Retor st was	ted te			density, wt pct	density. 15/ft3	wet density, 15/ft3	10	settlement pct	. 28 day	120 day	180 day	7 day	28 day	120 day	180 day
													TAIL INGS	0								
1	l	4	0	ŷ		0	ð	0	5.2	5.5	8	NA	113		0 3.	1 20	23	24	44	61	180	
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	3	1	0	0		0	0	0	6.3 7 F	5.0	: B() NA	119	1.52		b 23	23	29	100	130	186	
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	3	9	0	0)	0	0	0	2.5	2.5	. 8	4 N/	123		0 2.	1 71	89	83	171	215	238	
	8	3	0	C)	0	y .	0	3	:	5 8	2 NH	125	1.12	5 4,	2 45	5 69	69	174	215	280	
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	7	1.5	1.5		0	ò	ů	0	4	5.3	3 8	2 N	A 127	7 0	.5 3	.1 3	7 4	5 55	145	202	267	
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i	9	6	2		0	9	0	0	2.5	3.3	5 8	14 N	A 130)	0 4	.2 7	9 11-	4 135	286	393	556	
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	27	3	3		õ	0	0	0	- i		8	32 1	IA 12	6 0	.5 3	.1 2	3 4	ь <u>5</u> 0	109	131	277	7
	29	2	2		0	0	0	ð	6	1	12	91)	(A 12	S 0.	75 3	.8	0	0 0	77	94	109	,
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	30	4	0		0	0	0	0	8.2	8	.2	76 1	NA 12	0	7 3	.1 1	16 1	9 17	39	5	i 5()
	31	6	0		0	0	0	0	5.6	5	.6	76	NA 11	7 7.	.25	1	28 2	5 33	63	5 100) 90)
	32	8	0		0	0	Ů	0	4.3	4	.3	76 1	NA 11	87.	.25	1	5 4	8 51	91	124	9 13/	5
	33	10	0		0	0	0	0	3.0	5 5	.5	76	NG 12	11 in	/.3 () =	5.5	50 C	4 60	14.	5 I9:	1 240 7 200) D
	34 75	12	1 5		0	0	0	0	5.6	,	.7 68	76	NH 14 NA 11	.u .	(•J 4 75 7	(. (1	98 9 15 5	1 52	17. . A	L 24.	3 27" 7 11"	7 र
	34 74	۳.J ۶	1	•	ů n	5	ő	ô	4.3	5.	++ k9	76	NA II	9 9	.25	1.5	25	51 47	· •	3 12	/ 11. ñ 17	7
	37	7.5	2.5	5	õ	ò	0	0	3.5	4.	63	76	NA 12	20	9.5	4.1	52	57 74	11	4 17	9 25	C
	38	9	3		0	0	0	0	2.95	3.	93	76	NA E	21	10	4.2	NA I	ig Na	13	2 20	7 31	7
	39	3	1	5	0	0	0	0	5,58	11.	15	76	NA 13	22	11	6.2	13	6 10) 4	6 5	4 8	3
	40	2	3	2	0	0	Û	Û	8.21	16.	42	76	NA 11	21	11	6.6	0	0 0) 3	2 3	9 4	7
	41	i	;	3	0	0	0	0	8.21	32.	85	76	NA 13	23	11	7.3	0	0 0	2	7 3	7 5	2
	42	<u>6</u>	1	6	0	0	0	9	2.92		1.9	/6	NA 1	25	11	6.Z	25	57 62	28	1 12	9 24 F	-1
	43	3.4	1	U A	Ú A	Ų A	0.5	Q Q	4,4	4. E	.87	80 GA	лн 11 ма 11	/) 01 0	1	3.1 1 7	20	0/ 42 ⊀a 47	ז כ זי ד	8 11 7 0	.a 13 n 44	.) 7
	49 45	4.8 0		v 5	v A	0	5	ů ů	7.42	5,	142 NA	80	ля 1. NA 1	27 2 75 7	. 23	7+/ 5.7	0	งา 16 กิ 18	, b R	0 E	14. II NG 14.	.J .5
	46	ů.		0	0	ő	0	14	2.04		NA	80	NA 1	22 7	.75	5.2	0	0 14	A.	0 7	(9 4	
	47	ñ		0	0	0	ů.	28	1.14		NA	80	NG 1	· 77	2.5	5.5	11	0 N	4	A 10		10

NA Not available.

W/B Ratio of water-to-cement, fly ash, kiln dust, and oil shale retorted waste.

W/C Water-to-cement ratio.

r .

APPENDIX B.-STATISTICAL DEFINITIONS FOR GOODNESS-OF-FIT

<u>Linear Correlation Coefficient</u>.-Given a set of data (x_i, y_i) , the linear correlation coefficient, r, is defined by

$$r = \frac{\sum_{i=1}^{n} (x_{i} - \bar{x})(y_{i} - \bar{y})}{\sqrt{\sum_{i=1}^{n} (x_{i} - \bar{x})^{2} \sum_{i=1}^{n} (y_{i} - \bar{y})^{2}}},$$

where \overline{x} and \overline{y} are the respective means of the input data.

<u>Multiple Correlation Coefficient</u>.-As stated in reference 15, when using a multiple-regression model, there exists one dependent and 'p' predictor variables. The multiple correlation coefficient, R, measures the percent of variation accounted for by the model. It is defined by

$$\mathbf{R}^{2} = \frac{\sum_{i=1}^{n} (\tilde{y}_{i} - \bar{y})^{2}}{\sum_{i=1}^{n} (y_{i} - \bar{y})^{2}},$$

where $y_i =$ dependent variable,

 \tilde{y}_i = fitted (derived) dependent variable,

and \overline{y} = sample mean of y_i .

If there is only one predictor variable, R reduces to r. An R-value of 1 indicates that the model provides all necessary information; R = 0 implies that the p-dimensional hyperplane is an inadequate model.

Index of Determination.-The index of determination, I, is defined by

$$I = 1 - \frac{\sum_{i=1}^{n} (y_i - \tilde{y}_i)^2}{\sum_{i=1}^{n} (y_i - \bar{y})^2}$$

If the model being tested is linear, I is equivalent to either r or R. The index is useful in that it can be used to compare goodness-of-fit of nonlinear models.

APPENDIX C.-LINEAR REGRESSION RESULTS FOR TAILINGS A

The following abbreviations are used in this appendix:

7DCOMP	7-day compressive strength
CEMENT	Cement content, percent of tailings
Coef	Coefficient of variation
DF	Degrees of freedom
MS	Mean squares
NAp	Not applicable
Obs	Observation
R-sq	R^2 (multivariate correlation coefficient)
R-sq(adj)	R ₂ adjusted for degrees of freedom
\$	Estimated standard deviation about the regression line
SEQ SS	Sequential sum of squares
SS	Sum of squares
Stdev	Standard deviation
Stdev fit	Standard deviation of fitted value
St resid	Standardized residual
t-ratio	Coefficient/standard deviation
W/C	Water-to-cement ratio
-	

The regression equation is 7DCOMP = -73.6 + 32.4 CEMENT - 1.40 W/C. Of the 95 observations, only 90 were used; the remaining five observations contained missing values.

Predicto	r	Coef	Stdev	<u>t-ratio</u>
Constant CEMENT W/C		-73.64 32.356 -1.396	59.73 2.672 8.729	-1.23 12.11 16
s = 69.42	\mathbf{R} -sq = 87.9	9 pct	R-sq(adj) =	= 87.6 pct

Analysis of Variance

Source	DF	<u>SS</u>	<u>MS</u>	
Regression	2 87	3,032,842 419,299	1,516,421 4,820	
Total	89	3,452,140	NAp	
Source	DF	SEQ SS		
CEMENT W/C	1 1	3,032,719 123		

Obs.	CEMENT	7DCOMP	Fit	Stdev fit	Residual	St resid
1	4.0	48.00	44.20	26.66	3.80	0.06X
2	6.0	70.00	112.68	10.83	-42.68	62
3	8.0	112.00	179.35	8.41	-67.35	98
4	10.0	212.00	245.28	9.04	-33.28	48
5	12.0	278.00	310.76	9.27	-32.76	48
<u>6</u> . , <i>.</i>	14.0	340.00	376.03	9.53	-36.03	52
7	4.0	79.00	48.11	11.33	30.89	.45
8	4.0	48.00	47,41	12.03	.59	.01
9	4.0	54.00	46.71	14.12	7.29	.11
10	6.0	196.00	115.61	16.17	80.39	1.19
11	6.0	141.00	114.91	12.89	26.09	.38
12	6.0	118.00	114.22	10.44	3.78	.06
13	4.0	25.00	45.41	20.05	-20.41	31
14	7.0	106.00	145.95	8.91	-39.95	58
15	9.0	171.00	212.39	8.73	-41.39	60
10	12.0	274.00	2/8.13	9.02	-4.13	00
17	13.0	431,00	343.00	9.77	87,47	1.27
10	5.0	114.00	79 02	15 50	39.90	.00
20	5.0	158.00	170.93	8 10	-21.07	.02
20	0.0	100.00	(13.0)	0.10	-21.07	01
21	10.0	259.00	245.32	9.20	13.68	.20
22	13.0	459.00	343.31	9.03	115.69	1.68
23	15.0	518.00	408.63	9.91	109.37	1.59
24	18.0	603.00	506.14	12.12	96.86	1.42
20	0.0	47.00	44.96	22.00	014	0.0V
20	4.0	95.00	119.00	23.02	-19.21	.03A
28	80	130.00	179 74	9.38	-40 74	20
29	11.0	183.00	277.91	8.69	-94.91	-1.38
30	13.0	160.00	343.35	9.16	-183.35	-2.66R
31	14.0	365.00	376.05	9.61	-11.05	- 16
32	0.0	0.00	*	*	*	*
33	4.0	72.00	45.41	20.05	26.59	.40
34	7.0	83.00	145.95	8.91	-62.95	91
35	9.0	135.00	212.39	8.73	-77.39	-1.12
36	11.0	308.00	278.13	9.52	29.87	.43
37	13.0	324.00	343.53	9.77	-19.53	28
38	16.0	428.00	441.10	10.30	-13.10	19
39	5.0	79.00	79.42	13.45	42	01
40	8.0	131.00	179.39	8.48	-48.39	70
41	11.0	295.00	277.91	8.69	17.09	.25
42	13.0	361.00	343.50	9.67	17.50	.25
43	16.0	529.00	441.16	10.37	87.84	1.28
44	19.0	772.00	538.63	13.35	233.37	3.43R
45	8.0	129.00	179.39	8.48	-50.39	73
46	12.0	270.00	310.76	9.27	-40.76	59
47	16.0	457.00	441.16	10.37	15.84	.23
48	20.0	497.00	571.16	14.70	-74.16	-1.09
49	24.0	744.00	700.97	21.62	43.03	.65
50	28.0	803.00	830.68	29.88	-27.68	44X
51	0.0	0.00	*	*	*	*
52	5.0	88.00	78.93	15.58	9.07	.13
53	8.0	154.00	179.07	8.10	-25.07	36
54	10.0	264.00	245.32	9.20	18.68	.27
55	13.0	341.00	343.31	9.03	-2.31	03
56	15.0	501.00	408.63	9.91	92.37	1.34
57	18.0	357.00	506.14	12.12	-149.14	-2.18R
58	5.0	152.00	78.93	15.58	73.07	1.08
59	0.8	80.00	1/9.0/	8.10	-99.07	-1.44
UU	10.0	200. UU	240.32	9.20	- 12.32	18

Obs.	CEMENT	7DCOMP	Fit	Stdev fit	Residual	St resid
61	13.0	363.00	343.31	9.03	19.69	0.29
62	15.0	492.00	408.63	9.91	83.37	1.21
63	18.0	623.00	506.14	12.12	116.86	1.71
64	0.0	0.00	*	*	*	*
65	4.0	71.00	45.41	20.05	25.59	.38
66	7.0	134.00	145.95	8.91	-11.95	17
67	9.0	179.00	212.39	8.73	-33.39	48
68	11.0	307.00	278.13	9.52	28.87	.42
69	13.0	292.00	343.53	9.77	-51.53	75
70	16.0	451.00	441.10	10.30	9.90	.14
71	4.0	130.00	48.81	12.25	81.19	1.19
72	4.0	113.00	48.11	11.33	64.89	.95
73	4.0	88.00	47.41	12.03	40.59	.59
74	7.0	215.00	147.97	13.85	67.03	.99
75	7.0	137.00	147.27	10.81	-10.27	15
76	7.0	103.00	146.57	8.95	-43.57	63
77	0.0	0.00	*	*	*	*
78	5.0	89.00	79.42	13.45	9.58	.14
79	8.0	140.00	179.39	8.48	-39.39	57
80	11.0	441.00	277.97	8.89	163.03	2.37R
81	13.0	325.00	343.53	9.77	-18.53	27
82	16.0	466.00	441.19	10.41	24.81	.36
83	19.0	668.00	538.66	13.32	129.34	1.90
84	4.0	107.00	48.81	12.25	58.19	.85
85	4.0	99.00	48.11	11.33	50.89	.74
86	4.0	69.00	47.41	12.03	21.59	.32
87	8.0	149.00	180.33	11.68	-31.33	46
38	8.0	122.00	179.63	9.04	-57.63	- 84
B9	8.0	80.00	178.93	8.08	-98.93	-1.43
90	8.0	125.00	179.45	8.59	-54.45	79
91	12.0	216.00	310.80	9.42	-94.80	-1.38
92	16.0	322.00	441.19	10.41	-119.19	-1.74
93	20.0	420.00	571.18	14.66	-151.18	-2.23R
94	24.0	610.00	700.99	21.58	-90.99	-1.38
95	28.0	824.00	830.69	29.83	-66.69	11X

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* Calculation not possible. Zero value involved.
R Denotes observation with large standardized residual.
X Denotes observation whose X value gives it large influence.

APPENDIX D.-MATHEMATICAL REPRESENTATIONS AND INDICES OF DETERMINATION FOR EXPONENTIAL CURVES RELATING COMPRESSIVE STRENGTH TO WATER-TO-CEMENT RATIO

Index of determination, I

Tailings A, B, and C:	
$7DCOMP = 1,541.04e^{-0.58 W/C}$. 0.796
$28DCOMP = 2.782.30e^{-0.64 \text{ W/C}}$	764
120CDOMP = 3.690.00e ^{-0.58} W/C	658
$180DCOMP = 1.035.80e^{-0.25} W/C$	050
100DCOMI = 1,055.000	, ,410
Tailings A total	
Tamings A, total. TDCOMP 1 515 52-0.57 W/C	000
$DCOMP = 1,515.53e^{-10.62} W/C$	889
$28DCOMP = 2,836.1/e^{-0.52} W/C$	843
$120DCOMP = 4,008.05e^{-0.55}$	711
Tailings B, total:	
$7DCOMP = 2,632.73e^{-0.79} W/C \dots \dots$	944
$28DCOMP = 3,734.56e^{-0.79} \text{ w/C}$	893
$120DCOMP = 2,969.06e^{-0.57 \text{ W/C}}$	866
$180DCOMP = 1.587.58e^{-0.35 W/C}$	632
,,,	
Tailings C total	
$7DCOMP = 610.85e^{-0.29} W/C$	343
$28DCOMP = 702.260^{-0.27} W/C$	·
20DCOMF = 792.50C	203
$120DCOMP = 862.11e^{-10.21} W/C$	338
$180DCOMP = 891.91e^{-0.21 W/C} \dots$	312
Tailings A, no additives:	0.7.0
$7DCOMP = 1,106.35e^{-0.50 W/C} \dots$	982
$28DCOMP = 1,812.46e^{-0.54} \text{ w/C} \dots \dots$	975
$120DCOMP = 1,940.58e^{-0.50 \text{ w/C}}$	976
Tailings B, no additives:	
$7DCOMP = 2.692.67e^{-0.79 W/C}$	956
$28DCOMP = 4.392.84e^{-0.88} W/C$	920
$120DCOMP = 4.316.10e^{-0.77} W/C$	
$120DCOMP = 1.772.272^{-0.50} W/C$,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,
180DCOWP = 1,72.576	940
Tailings C. no additives	
$7D_{COMP}$ 1.252.770.65 W/C	000
/DCOMP = 1,352.7/6	892
$28DCOMP = 2,004.24e^{-0.05 W/C}$	879
$120DCOMP = 1,832.01e^{-0.56 W/C} \dots \dots$	839
$180DCOMP = 1,860.52e^{-0.58 \text{ W/C}} \dots \dots$	904
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DCOMP 7-, 28-, 120-, or 180-day compressive strength. W/C Water-to-cement ratio.

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