# Cemented Rockfill QA/QC Batching Study with Specific Focus on Different Sample Preparation Techniques

Tyler Emery<sup>1</sup>, Josef Bourgeois<sup>1</sup>, Brad Seymour<sup>1</sup>, David Sweet<sup>1</sup>, David Porter<sup>2</sup>

<sup>1</sup>National Institute for Occupational Safety and Health, Spokane, WA, USA, qdn0@cdc.gov <sup>2</sup>Lundin Mining, Michigamme, MI, USA, david.porter@lundinmining.com

#### **Abstract**

Cemented rockfill (CRF) is being used to backfill primary and secondary stopes at the Eagle Mine, an underground nickel and copper mine operated by Lundin Mining. The National Institute for Occupational Safety and Health (NIOSH) Spokane Mining Research Division (SMRD) is partnering with Eagle Mine to research the advancement of mine backfill QA/QC guidelines through batching and strength testing method studies, specifically through investigation of existing methods that best determine *in situ* properties of CRF. As part of this collaborative research, NIOSH researchers traveled to Eagle Mine and worked with mine staff to trial multiple 6 in QA/QC CRF cylinder preparation methods including ASTM C31, ASTM C1435, and a drop hammer compaction method to identify which method can best correlate to the *in situ* strength of CRF placed underground with the least operator bias and highest confidence. A total of 60 cylindrical samples (6 in) of CRF were cast at the mine and cured underground, in accordance with the three-cylinder preparation methods,. Samples were later transported to the SMRD laboratory to determine the strength and elastic properties of the CRF. Data collected from this study will help identify which test cylinder method provides consistent density and strength results that can better correlate with *in situ* density of CRF.

Key words: ASTM, cylinder preparation, backfill strength, underground mining

#### Introduction

Cemented rockfill (CRF) is a zero or low slump, coarse aggregate, concrete-like engineered material used for backfilling mined-out openings. The defining characteristics of CRF include: 1) large aggregate top size of 2 to 6 inches, 2) inclusion of fines from the crushing process, and 3) less than 1 in slump, which is essential for creating self-standing faces when jamming the material. According to Stone, aggregate gradation controls the density of the CRF, and therefore has a significant impact on CRF strength (Stone, 2007). When utilized in underground environments, backfill strength requirements are typically determined through a combination of analytical formulae (Mitchell 1991) and empirical design (Pakalnis, Caceres et al., 2005).

#### Previous cemented rockfill research at NIOSH

Research from Seymour et al. (2013) discovered that backfill failures in U.S. mines are usually attributed to inadequate backfill strength, insufficient or inconsistent quality control measures, or larger-than-expected mining spans. This 2013 study prompted researchers at the National Institute for Occupational Safety and Health (NIOSH) Spokane Mining Research Division (SMRD) to investigate the strength and mechanical properties of CRF used as part of the mining cycle at collaborating underhand cut-and-fill mining operations through backfill sample cylinder size effect studies (Warren, Raffaldi et al., 2018, Stone, Pakalnis and Seymour, 2019). Through this investigation, it was discovered that sample density of Quality Assurance/Quality Control (QA/QC) CRF cylinders has a significant effect on its resulting unconfined compressive strength (UCS). If there is a large variance in QA/QC cylinder density/strength due to the sampling procedure used and/or change in cylinder preparer, it can become difficult to confidently estimate strength properties of emplaced CRF in underground mine openings.

In 2021, a collaborative study between researchers at NIOSH and geotechnical engineers at Eagle Mine further investigated the correlation between CRF QA/QC cylinder density and strength (Bourgeois et al., 2023) with an additional research task of comparing the ASTM C31 and ASTM C1435 sample preparation techniques. A major conclusion of the study was that the ASTM C1435 sample preparation technique resulted in a more consistent density and UCS of QA/QC 6 × 12 in cylinders compared to that of ASTM C31, and that more data were needed to back up these initial findings. Therefore, this paper focuses on a follow-up study which provides more data points of the resulting strength properties of CRF QA/QC cylinders based on the currently used preparation techniques including ASTM C31, ASTM C1435, and a drop hammer compaction method.

## Cemented rockfill QA/QC cylinder preparation methods used in this study *ASTM C31*

The most commonly used standard for sampling CRF is ASTM C31 (2022): Standard practice for making and curing concrete test specimens in the field. For this study,  $6 \times 12$  in cylinders were molded by rodding three layers of approximately equal depth at 25 roddings per layer. The process of molding by rodding according to C31 is shown in Figure 1a.

### **ASTM C1435**

The second CRF sampling method used in this study was ASTM C1435 (2020): "Standard practice for molding roller-compacted concrete in cylinder molds using a vibrating hammer. This standard is more commonly used in civil engineering applications and was first explored by SMRD for use in CRF sampling in the Bourgeois et al. (2023) study. This method was investigated as the methodology and equipment use appeared to be a promising means of reducing variability in sample density due to operator bias. For this study,  $6 \times 12$  in cylinders were molded through the following steps:

- 1. Vibrating hammer was placed with tamping plate into the  $6 \times 12$ -inch cylinder mold onto the CRF.
- 2. Vibrating hammer was started and CRF was allowed to consolidate either until the mortar formed a ring around the total perimeter of the plate or until 15 secs had passed.
- 3. Process was repeated for three layers of approximately equal depth (similar to ASTM C31).
- 4.Mold was completed by overfilling it with CRF and consolidating with the vibrating hammer and tamping plate to create a smooth and level top.

The process of molding by vibrating hammer according to ASTM C1435 is shown in Figure 1b.

## Drop hammer compaction method

The final CRF method used in this study was a drop hammer method that was developed using a modified dynamic cone penetrometer which had a plat face plate in place of the cone, which was developed for use by Eagle Mine. The standard ASTM method D7380 for Soil Compaction Determination at Shallow Depths does not apply for this case, so through experience the mine had determined that 10 drops per lift (total of 3 lifts) to be adequate for proper compaction. The process of molding by the drop hammer compaction method is shown in Figure 1c).





Figure 1. Example of researcher constructing 6 × 12-inch CRF cylinder according to a) ASTM C31, b) ASTM C1435, and c) drop hammer compaction method.

## Cemented Rockfill QA/QC Cylinder Construction at Eagle Mine

The principal mining method at Eagle is longhole open stoping with delayed rockfill emplacement. A detailed description of the mining sequencing, CRF material composition, mixing, transportation, and quality control practices can be found in Bourgeois et al., (2023). In June 2023, researchers at SMRD and geotechnical staff from Eagle Mine constructed 60 of  $6 \times 12$  in cylindrical CRF cylinders in accordance with ASTM C31, ASTM C1435, and the drop hammer compaction method for strength testing at the SMRD lab in Spokane, WA.

CRF cylinder construction occurred underground at the -473 level stope of the Eagle workings where backfill was being poured and jammed into a mined-out stope that same day. The sixty constructed cylinders were made according to the three sample preparation methods in the following manner:

- •10 ASTM C31 (not-screened)
- •10 ASTM C31 (screened)
- •10 ASTM C1435 (not-screened)
- •10 ASTM C1435 (screened)
- •10 drop hammer (not-screened)
- •10 drop hammer (screened)

Screening refers to the removal of coarse aggregate larger than two inches in accordance with ASTM C31 (Figure 2), which requires removal of aggregate larger than one third the diameter of the test cylinder. In past NIOSH CRF studies, which primarily followed ASTM C31 for cylinder construction, wet-sieving for two inch plus aggregate was strictly followed when casting 6 × 12 in cylinders as some CRF mixes contained maximum aggregate sizes up to four inches (Stone, Pakalnis, and Seymour, 2019). In the case of the study outlined in this paper, with a maximum aggregate size of 3 in used at Eagle Mine, the research team wanted to see the effect that screening versus no-screening had on density and UCS for all sample preparation techniques. Additionally, the cylinder preparer's initials were noted according to each sample preparation method to capture the variable of preparer bias.



Figure 2. Process of screening two inch plus sized coarse aggregate for screened  $6 \times 12$  in CRF cylinders.

After the  $6 \times 12$  in CRF cylinders were constructed, they were left to cure at the -473 level stope underground for 15 days before being shipped to the SMRD lab for 28 day strength testing. A wooden container with foam inserts for secure cylinder placement was developed and bolted to a pallet for safe, undisturbed shipment of the CRF samples. Figure 3 shows samples after form removal.



Figure 3. Example of cured, not-screened samples after molds were removed a) ASTM C31, b) ASTM 1435, c) drop hammer compaction method.

## **Strength Testing Methodology**

Prior to strength testing, the CRF cylinders were stripped from their forms and final dimensions were measured to identify discrepancies from their diameter and length. Every CRF cylinder was weighed to account for variations in density and the presence of voids. While the plan was to test the  $6 \times 12$  in CRF cylinders at 28 day strength, the actual testing was delayed to 41 day strength due to extra time needed

for sample preparation, particularly in cutting the ends of every sample to make them level for UCS testing according to ASTM standard C39 (2021). A 200,000 lbs capacity Tinius Olsen test machine was used for the 41 day strength testing with steel platens placed at each end of the CRF cylinder.

## **Strength Testing Results**

UCS test results of each CRF cylinder constructed at the -473 level stope of Eagle Mine are summarized in Table 1. Average values of density and UCS, as well as coefficient of variation according to CRF cylinder preparation method is shown in Table 2. Figure 4 plots results of all 60 samples according to CRF cylinder preparation method, density, and resulting UCS. Figure 5 plots averages of the results shown in Figure 4.

Table 1. CRF cylinder measuring and UCS results according to compaction method and sampler.

Sample	<b>~</b>						i I
1 1	Compaction	Sampler	Height	Diameter	Weight	Density	UCS
ID	Method	ID	(in)	(in)	(lbs)	(lbs/ft³)	(psi)
C1*			11.604	6.025	27.07	145 1	261
S1*			11.604	6.035	27.87	145.1	361
S2		TDE	11.730	6.251	29.62	142.2	1,300
S3		TE	11.451	6.006	29.15	155.3	1,580
S4			11.663	5.988	30.01	157.9	1,736
S5			11.549	5.980	30.23	161.0	2,107
S6			11.479	5.986	29.98	160.4	2,231
S7			11.584	5.981	30.23	160.5	2,366
S8			11.448	5.964	29.14	157.4	1,874
S9			11.561	5.975	30.10	160.4	2,193
S10	ASTM C31	FR	11.610	5.987	29.62	156.6	1,881
US1	7 <b>15</b> 11 <b>11</b> C5 1	TIX	11.563	6.036	29.18	152.4	1,583
US2			11.043	6.038	27.92	152.6	1,705
US3			11.338	6.022	28.62	153.2	1,318
US4			10.960	6.032	28.63	158.0	2,014
US5			11.507	6.017	28.45	150.3	1,460
US6*			10.780	6.045	23.66	132.2	585
US7			11.817	6.023	28.22	144.8	1,004
US8		TE	11.508	6.046	28.13	147.1	1,479
US9			11.164	5.999	26.75	146.5	1,221
US10			11.673	6.020	29.67	154.3	1,664
S1	ASTM C1435		11.600	6.068	30.53	157.2	2,190
S2			11.573	6.034	30.33	158.4	1,984
S3		TE	11.631	6.066	30.73	158.0	2,107
S4			11.573	6.049	29.18	151.6	1,610
S5			11.485	6.053	28.97	151.5	1,582
S6			11.519	6.022	29.24	154.0	1,600
S7			11.433	6.014	30.13	160.3	2,253
S8		TL	11.587	6.023	30.41	159.2	2,167
S9			11.569	6.049	30.23	157.1	2,384
S10			11.578	5.995	30.32	160.3	2,223
US1		FR	11.566	6.060	30.51	158.0	1,890
US2			11.660	6.047	30.56	157.7	2,358
US3			11.492	6.014	30.48	161.4	2,262
US4			11.579	6.098	30.47	155.7	1,973

US6         US7         Interpretation of the property of the propert								
US7         US8         11.810         6.042         31.02         158.3         1,87           US9         11.652         6.037         30.51         158.1         2,02           US10         11.080         6.038         27.27         148.5         1,17           US10         11.698         6.052         30.53         156.8         1,80           S1         11.325         6.020         29.23         156.7         2,07           S2         11.533         6.028         29.46         154.7         1,61           S3         11.308         5.924         28.67         158.9         1,74           S4         11.430         6.023         29.56         156.8         2,29           S5         11.420         5.977         28.65         154.5         1,71           S6         11.492         6.050         29.40         153.8         1,85           S7         88         11.500         6.045         29.78         155.9         1,76           S9         11.537         6.075         29.76         153.8         1,83           US1         11.543         5.982         29.64         157.9         2,00	US5			11.373	6.103	28.86	149.9	1,437
US8         US9         BS         11.652         6.037         30.51         158.1         2,02           US10         11.080         6.038         27.27         148.5         1,17           US10         11.698         6.052         30.53         156.8         1,80           S1         11.325         6.020         29.23         156.7         2,07           S2         11.533         6.028         29.46         154.7         1,61           S3         11.430         6.023         29.56         156.8         2,29           S5         11.420         5.977         28.65         154.5         1,71           S6         11.492         6.050         29.40         153.8         1,85           S7         11.627         6.028         29.93         155.8         2,23           S8         11.500         6.045         29.78         155.9         1,76           US1         US1         11.543         5.982         29.64         157.9         2,00           US2         US3         11.718         6.042         29.39         151.2         1,70           US4         US4         11.327         6.033	US6		BS	11.589	6.022	29.76	155.8	1,361
US9         11.080         6.038         27.27         148.5         1,17           US10         11.698         6.052         30.53         156.8         1,80           S1         11.325         6.020         29.23         156.7         2,07           S2         11.533         6.028         29.46         154.7         1,61           S3         11.430         6.023         29.56         158.9         1,74           S4         11.430         6.023         29.56         156.8         2,29           S5         11.420         5.977         28.65         154.5         1,71           S6         11.492         6.050         29.40         153.8         1,85           S7         11.627         6.028         29.93         155.8         2,23           S8         11.500         6.045         29.78         155.9         1,76           US1         11.428         6.015         28.76         153.1         1,80           US2         11.718         6.042         29.39         151.2         1,70           US3         US4         11.327         6.033         29.34         153.6         2,09	US7			11.810	6.042	31.02	158.3	1,873
US10         11.698         6.052         30.53         156.8         1,80           S1         11.325         6.020         29.23         156.7         2,07           S2         11.533         6.028         29.46         154.7         1,61           S3         5.924         28.67         158.9         1,74           S4         11.430         6.023         29.56         156.8         2,29           S5         11.420         5.977         28.65         154.5         1,71           S6         11.492         6.050         29.40         153.8         1,85           S7         88         11.627         6.028         29.93         155.8         2,23           S8         11.500         6.045         29.78         155.9         1,76           S9         11.537         6.075         29.76         153.8         1,83           S10         US1         11.543         5.982         29.64         157.9         2,00           US2         11.718         6.042         29.39         151.2         1,70           US3         US4         11.327         6.033         29.34         153.6         2,09	US8			11.652	6.037	30.51	158.1	2,029
S1   S2   S2   S3   DP   11.325   6.020   29.23   156.7   2,07     S3   S4   S5   S6   S7   S8   S9   S10   US1   US2   US3   US4   US2   US4   US4   US4   US2   US4   US4   US2   US4   US4   US2   US4   U55   U56   U57   U58   U58	US9			11.080	6.038	27.27	148.5	1,175
S2     DP     11.533     6.028     29.46     154.7     1,61       S3     11.308     5.924     28.67     158.9     1,74       S4     11.430     6.023     29.56     156.8     2,29       S5     11.420     5.977     28.65     154.5     1,71       S6     11.492     6.050     29.40     153.8     1,85       S7     11.627     6.028     29.93     155.8     2,23       S9     11.500     6.045     29.78     155.9     1,76       11.537     6.075     29.76     153.8     1,80       US1     11.543     5.982     29.64     157.9     2,00       US2     11.718     6.042     29.39     151.2     1,70       US3     US4     11.550     6.033     29.34     153.6     2,09       11.327     6.033     28.49     152.0     1,58	US10	1		11.698	6.052	30.53	156.8	1,808
S3     DP     11.308     5.924     28.67     158.9     1,74       S4     11.430     6.023     29.56     156.8     2,29       S5     11.420     5.977     28.65     154.5     1,71       S6     11.492     6.050     29.40     153.8     1,85       S7     11.627     6.028     29.93     155.8     2,23       S8     11.500     6.045     29.78     155.9     1,76       S9     11.537     6.075     29.76     153.8     1,80       US1     11.428     6.015     28.76     153.1     1,80       US2     11.718     6.042     29.39     151.2     1,70       US3     US4     11.550     6.033     29.34     153.6     2,09       US4     11.327     6.033     28.49     152.0     1,58	S1			11.325	6.020	29.23	156.7	2,078
S4     11.430     6.023     29.56     156.8     2,29       S5     11.420     5.977     28.65     154.5     1,71       S6     11.492     6.050     29.40     153.8     1,85       S7     11.627     6.028     29.93     155.8     2,23       S8     11.500     6.045     29.78     155.9     1,76       S9     11.537     6.075     29.76     153.8     1,83       11.428     6.015     28.76     153.1     1,80       US1     11.543     5.982     29.64     157.9     2,00       US2     11.718     6.042     29.39     151.2     1,70       US3     US4     11.550     6.033     29.34     153.6     2,09       11.327     6.033     28.49     152.0     1,58	S2	1	DP	11.533	6.028	29.46	154.7	1,611
S5     11.420     5.977     28.65     154.5     1,71       S6     11.492     6.050     29.40     153.8     1,85       S7     11.627     6.028     29.93     155.8     2,23       S8     11.500     6.045     29.78     155.9     1,76       S10     11.537     6.075     29.76     153.8     1,83       US1     11.543     5.982     29.64     157.9     2,00       US2     11.718     6.042     29.39     151.2     1,70       US3     US4     11.550     6.033     29.34     153.6     2,09       11.327     6.033     28.49     152.0     1,58	S3			11.308	5.924	28.67	158.9	1,747
S6         11.492         6.050         29.40         153.8         1,85           S7         11.627         6.028         29.93         155.8         2,23           S8         11.500         6.045         29.78         155.9         1,76           S9         11.537         6.075         29.76         153.8         1,83           US1         11.428         6.015         28.76         153.1         1,80           US2         11.543         5.982         29.64         157.9         2,00           US3         11.718         6.042         29.39         151.2         1,70           US4         11.550         6.033         29.34         153.6         2,09           11.327         6.033         28.49         152.0         1,58	S4			11.430	6.023	29.56	156.8	2,297
S7         S8         11.627         6.028         29.93         155.8         2,23           S9         11.500         6.045         29.78         155.9         1,76           S10         11.537         6.075         29.76         153.8         1,83           11.428         6.015         28.76         153.1         1,80           US1         11.543         5.982         29.64         157.9         2,00           11.718         6.042         29.39         151.2         1,70           US3         11.550         6.033         29.34         153.6         2,09           11.327         6.033         28.49         152.0         1,58	S5			11.420	5.977	28.65	154.5	1,713
S8         BS         11.500         6.045         29.78         155.9         1,76           S9         11.537         6.075         29.76         153.8         1,83           S10         11.428         6.015         28.76         153.1         1,80           US1         11.543         5.982         29.64         157.9         2,00           US2         11.718         6.042         29.39         151.2         1,70           US3         11.550         6.033         29.34         153.6         2,09           US4         11.327         6.033         28.49         152.0         1,58	S6			11.492	6.050	29.40	153.8	1,853
S9         11.537         6.075         29.76         153.8         1,83           US1         US1         11.428         6.015         28.76         153.1         1,80           US2         11.543         5.982         29.64         157.9         2,00           11.718         6.042         29.39         151.2         1,70           US3         11.550         6.033         29.34         153.6         2,09           11.327         6.033         28.49         152.0         1,58	S7		BS	11.627	6.028	29.93	155.8	2,233
S10         Drop Hammer         11.428         6.015         28.76         153.1         1,80           US1         11.543         5.982         29.64         157.9         2,00           US2         11.718         6.042         29.39         151.2         1,70           US3         11.550         6.033         29.34         153.6         2,09           US4         11.327         6.033         28.49         152.0         1,58	S8			11.500	6.045	29.78	155.9	1,763
US1         Drop Hammer         11.543         5.982         29.64         157.9         2,00           US2         11.718         6.042         29.39         151.2         1,70           US3         US4         11.550         6.033         29.34         153.6         2,09           11.327         6.033         28.49         152.0         1,58	S9			11.537	6.075	29.76	153.8	1,831
US2 US3 US4 US4 TE 11.543 5.982 29.64 157.9 2,00 11.718 6.042 29.39 151.2 1,70 11.550 6.033 29.34 153.6 2,09 11.327 6.033 28.49 152.0 1,58	S10			11.428	6.015	28.76	153.1	1,805
US3     TE     11.550     6.033     29.34     153.6     2,09       US4     11.327     6.033     28.49     152.0     1,58	US1		TE	11.543	5.982	29.64	157.9	2,006
US4 11.327 6.033 28.49 152.0 1,58	US2			11.718	6.042	29.39	151.2	1,704
	US3			11.550	6.033	29.34	153.6	2,099
11 (12   5 0(1   20 (7   150 2   1 01	US4			11.327	6.033	28.49	152.0	1,583
US3   11.612   5.961   29.67   158.2   1,91	US5			11.612	5.961	29.67	158.2	1,912
US6 11.506 6.018 29.47 155.6 1,75	US6		TL	11.506	6.018	29.47	155.6	1,756
US7* 11.483 5.942 27.02 146.6 875	US7*			11.483	5.942	27.02	146.6	875
US8 TL 11.051 5.954 28.37 159.3 2,00	US8			11.051	5.954	28.37	159.3	2,009
US9 11.721 5.877 29.75 161.7 1,68	US9			11.721	5.877	29.75	161.7	1,688
US10 11.553 6.004 29.22 154.4 1,54				11.553	6.004			1,548

Note: 'S' stands for screened, 'US' stands for un-screened, '\*' stands for anomalous test result which was discarded and not included in average values

Table 2. Average results excluding the three anomalous tests noted in Table 1.

Compaction	Screened or	Average Density	Average UCS	Coefficient of
Method	Un-Screened	$(lbs/ft^3)$	(psi)	Variation (%)
ACTM C21	Screened	156.9	1,919	17.9
ASTM C31	Un-Screened	151.0	1,494	19.8
ASTM C1435	Screened	156.8	2,010	15.0
ASTIVI C1433	Un-Screened	156.0	1,817	21.2
Окан Цаттак	Screened	155.4	1,893	12.2
Drop Hammer	Un-Screened	156.0	1,812	11.1

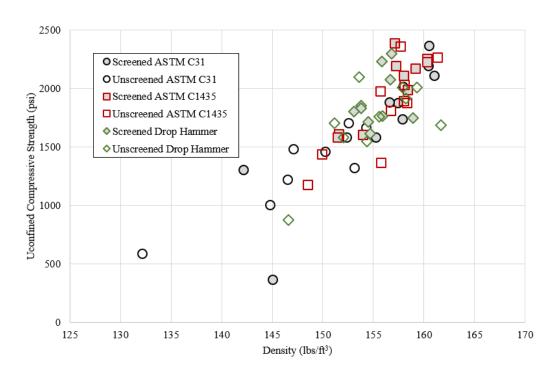


Figure 4. Unconfined compressive strength versus density of Eagle Mine CRF compaction study samples according to sampling method (n = 60).

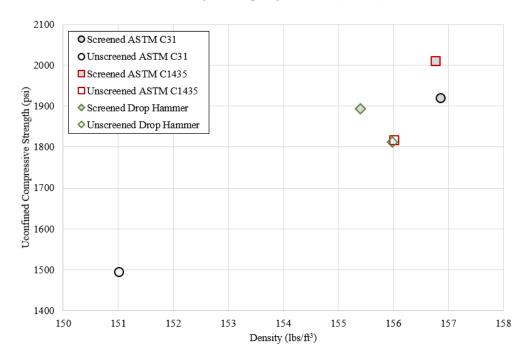


Figure 5. Average unconfined compressive strength versus density of Eagle Mine CRF compaction study samples according to sampling method excluding the three anomalous tests noted in Table 1 (n = 57).

The drop hammer compaction method demonstrated the least amount of variation in resulting UCS across all tested samples (Table 3). Additionally, there was minimal difference in average UCS when comparing the screened vs not-screened samples that were constructed using the drop hammer

compaction method. Table 3 shows variation results specific to the sample preparer, further outlining how changing preparers for the same cylinder construction method can have a significant effect on the consistency of the UCS test. Interestingly, while the drop hammer compaction method resulted with the most consistent UCS results, it also resulted in average UCS values similar to that of the ASTM C31 screened samples, which is commonly used in the mining industry and past NIOSH studies. The difference in average UCS between the drop hammer screened method and ASTM C31 screened is 26 PSI while the difference between ASTM C1435 screened and ASTM C31 screened is 91 PSI. The ASTM 1435 was designed to achieve maximum density of cylinder compaction, so this may not be applicable if a particular mining operation achieves less than the maximum compaction when placing CRF.

Table 3. Coefficient of variation in resulting UCS according to sample preparer.

Compaction	Sampler	Screened or Un-	Average UCS	Coefficient of Variation
Method	ID	Screened	(psi)	(%)
	TE	Screened	1,681	20.0
ASTM C31	FR	Screened	2,109	10.5
ASTIVI CS1	TE	Un-Screened	1,342	21.6
	FR	Un-Screened	1,616	16.4
	TE	Screened	1,895	14.9
ASTM C1435	TL	Screened	2,125	14.3
ASTWIC1433	FR	Un-Screened	1,984	18.3
	BS	Un-Screened	1,649	22.0
	DP	Screened	1,616	16.4
Duon Hommon	BS	Screened	1,897	10.1
Drop Hammer	TL	Un-Screened	1,750	11.0
	TE	Un-Screened	1,861	11.5

#### **Conclusions**

Based on testing data it was observed that:

- •The modified drop hammer compaction test produced the most consistent test cylinders.
- •The modified drop hammer method also produced the least variation in testing result.
- •Molding cylinders per ASTM 1435 with a compaction hammer also produced more consistent test cylinders and less variation between individuals casting cylinders.
- •The ASTM 1435 method is designed for achieving maximum density for the test cylinder and may not be applicable to operations that are employee placement techniques that do not achieve maximum density.
- •ASTM 1435 cylinders need to be cast with proper support for the plastic cylinder molds or they may deform.
- •Using ASTM C31 for unscreened material (backfill mix with maximum aggregate > 2 in) resulted in much lower UCS average value.
- •ASTM C31 also produced one of the two highest and the single lowest UCS values tested.

NIOSH will continue researching additional methods applicable to the mining industry and based on this research may investigate these methods further. More work needs to be done to develop and mature the drop hammer test method before it could be adopted, such as the best or most available apparatus for the drop hammer and a standard test method for casting cylinders. The ASTM 1435 method also shows promise for potential use as a standard method or perhaps this method should be modified. Finally, more investigation could be done to see if limiting the time for running the compaction hammer can be controlled to limit the amount of compaction and create more consistent results.

## **Declaration of Competing Interest**

The authors declare that they have no known competing financial interests or personal relationships that could have appeared to influence the work reported in this paper.

#### **Disclaimer**

The findings and conclusions in this report are those of the author(s) and do not necessarily represent the official position of the National Institute for Occupational Safety and Health, Centers for Disease Control and Prevention. Mention of any company or product does not constitute endorsement by NIOSH.

#### References

- ASTM (2020). "Standard practice for molding roller-compacted concrete in cylinder molds using a vibrating hammer." C1435/C1435M-20.
- ASTM (2021). "Standard test method for compressive strength of cylindrical concrete specimens." C39/C39M-21.
- ASTM (2022). "Standard practice for making and curing concrete test specimens in the field." C31/C31M-22.
- Bourgeois, J., T. Emery, B. Seymour, D. Sweet and D. Porter (2023). Cemented rockfill size effect study with specific focus on different sample preparation techniques. <a href="Mailto:SME Annual Meeting">SME Annual Meeting</a>. Denver, CO, Society for Mining, Metallurgy, & Exploration.
- Mitchell, R. J. (1991). "Sill mat evaluation using centrifuge models." Mining Science and Technology 13(3): 301-313.
- Pakalnis, R., C. Caceres, K. Clapp, M. Morin, T. Brady, T. Williams, W. Blake and M. MacLaughlin (2005). Design spans underhand cut and fill mining. <u>107th CIM-AGM</u>. Toronto, Canada, Canadian Institute of Mining, Metallurgy and Petroleum.
- Seymour, J. B., L. A. Martin, C. C. Clark, D. R. Tesarik and M. A. Stepan (2013). An analysis of recent MSHA accident data for underground metal mines using backfill. <a href="SME Annual Meeting">SME Annual Meeting</a>. Denver, CO, Society for Mining, Metallurgy, & Exploration.
- Stone, D. (2007). Factors that affect cemented rockfill quality in Nevada mines. Minefill 2007. Montreal, Quebec, Canadian Institute of Mining and Metallurgy.
- Stone, D., R. Pakalnis and B. Seymour (2019). Interpreting backfill QA/QC test data: Do we need an industry standard? <u>SME Annual Meeting</u>. Denver, CO, Society for Mining, Metallurgy, & Exploration.
- Warren, S. N., M. J. Raffaldi, K. K. Dehn, J. B. Seymour, L. A. Sandbak and J. Armstrong (2018). Estimating the unconfined compressive strength (UCS) of emplaced cemented rockfill (CRF) from QA/QC cylinder strengths. SME Annual Meeting. Minneapolis, MN, Society for Mining, Metallurgy, & Exploration.
- Warren, S. N., M. J. Raffaldi, K. K. Dehn, J. B. Seymour, L. A. Sandbak, J. Armstrong and M. Ferster (2018). Estimating the strength and mechanical properties of cemented rockfill for underhand cut-and-fill mines. 52nd US Rock Mechanics / Geomechanics Symposium. Seattle, WA, American Rock Mechanics Association.