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# Cold crucible induction melter test for crystalline ceramic waste form fabrication: A feasibility assessment



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# ABSTRACT

The first scaled proof-of-principle cold crucible induction melter (CCIM) test to process a multiphase ceramic waste form from a simulated combined (Cs/Sr, lanthanide and transition metal fission products) commercial used nuclear fuel waste stream was recently conducted in the United States. X-ray diffraction, 2-D X-ray absorption near edge structure (XANES), electron microscopy, inductively coupled plasma-atomic emission spectroscopy (and inductively coupled plasma-mass spectroscopy for Cs), and product consistency tests were used to characterize the fabricated CCIM material. Characterization analyses confirmed that a crystalline ceramic with a desirable phase assemblage was produced from a melt using a CCIM. Primary hollandite, pyrochlore/zirconolite, and perovskite phases were identified in addition to minor phases rich in Fe, Al, or Cs. The material produced in the CCIM was chemically homogeneous and displayed a uniform phase assemblage with acceptable aqueous chemical durability. © 2017 Elsevier B.V. All rights reserved.

# 1. Introduction

Reprocessing commercial used nuclear fuel (UNF) generates a waste stream that is commonly immobilized in borosilicate glasses because these glasses have demonstrated the capacity to accommodate a wide range of fission products and the technologies employed for processing glass are well developed. The United States continues to evaluate commercial UNF reprocessing through research conducted under the DOE-NE Fuel Cycle Technologies Program. As part of the that program, alternative waste forms are being investigated for the purpose of immobilizing projected waste streams generated from future reprocessing of commercial UNF. The primary waste forms of interest are glass-ceramic composites and ceramics. The former allows for the incorporation of fission products with limited solubility in borosilicate glass by intentional crystallization into chemically durable phases such as oxyapatite,

\* Corresponding author. E-mail address: jake.amoroso@srnl.doe.gov (J.W. Amoroso). powellite, lanthanide-borosilicate, and cerianite [1,2]. The principal advantage of ceramic waste forms is that waste ions are incorporated (high-level waste loadings vary from 10 to 35 wt%) into durable mineral phases, which are considerably less soluble in water relative to vitreous silicates, especially above atmospheric temperatures and pressures. High temperatures and pressures may be of concern because natural intrusive water will eventually enter the waste storage and disposal area over long time periods. In such a scenario, the decay heat of the waste form builds natural convective flow currents for the intrusive water. This heated water, after leaching from the given waste form, flows away, cools, precipitates the leached material, and then is recirculated back to the waste form, thereby restarting the leaching cycle.Many standard durability tests do not take into account such thermal conditions but, when considered, non-crystalline waste forms (including glass) are not as durable under hydrothermal conditions [3-6].

The most widely studied ceramic waste forms are derived from Synroc (i.e., synthetic rock) materials developed in Australia in the late 70's [7]. Synroc-type phases are composed of geochemically stable titanate minerals and are attractive for high-level waste (HLW) immobilization due to their ability to incorporate nearly all elements present in HLW, including fission products and minor actinides, in a crystalline lattice. The stability of synroc, and similar man-made analogs that mimic naturally occurring minerals, is substantiated by geological specimens containing uranium, thorium, and other naturally occurring radioactive isotopes recovered from the earth's crust after millions of years in the natural environment.

Ceramic formulations are generally developed to target Bahollandite (BaA12Ti6O16), zirconolite (CaZrTi2O7), and perovskite (CaTiO<sub>3</sub>), which are the characteristic titanate minerals comprising synroc. Elements with a  $3 + \text{ or } 2 + \text{ valance form perovskite-} ((A^{+2}))$ TiO<sub>3</sub>) and pyrochlore  $((A^{+3})_2Ti_2O_7)$ -type phases [8–9], while zirconium (4 + valence) partitions to a zirconolite (CaZrTi<sub>2</sub>O<sub>7</sub>) phase [10]. Cs and Rb elements partition to a hollandite structure based on the general formula  $Ba_x Cs_y M_z Ti^{+4}_{8-z} O_{16}$ , where z = 2x + y for trivalent cations and z = x + y/2 for divalent cations in order to maintain charge neutrality [11–13]. Zirconolite and pyrochlore are the major immobilization hosts for actinides, such as Pu, and the rare earth elements (i.e., Sc, Y, and the lanthanides), whereas perovskite is the principal immobilization host for Sr. Ba-hollandite is primarily used to immobilize Cs, K, Rb, and Ba. Depending on the waste composition, other minor synthetic Synroc mineral phases can be included, such as other forms of hollandite, magnetoplumbite (also for Sr), pyrochlore, and rutile (TiO<sub>2</sub>).[3]

Synroc materials are typically processed using some type of solid-state reaction, most notably hot isostatic pressing (HIP), which has been extensively developed by the Australian Nuclear Science and Technology Organization (ANSTO). However, this type of bulk ceramic processing in radioactive environments can be complex and has limited the use of ceramics for waste immobilization. As an alternative, synroc-type materials can be fabricated through a melt process, where crystallization and solidification occur upon cooling from the melt temperature [14–16]. In this manner, the bulk processing of ceramic forms would be simplified, and the advantages of a crystalline ceramic waste form can be combined with both existing processing flowsheets and knowledge currently implemented in waste vitrification processes.

Savannah River National Laboratory (SRNL) is developing a melt-process to convert waste generated from UNF reprocessing into a durable ceramic waste form as part of the United States DOE-NE Fuel Cycle Technologies Program [17]. The waste form is designed to crystallize into an engineered multiphase synroc-type ceramic upon cooling from a melt (melt processing). However, the relatively high melting temperatures (>1500 °C) of synroc-type ceramics (as well as glass-ceramics) limits the number of practical melter designs and excludes the use of traditional joule heated melters (JHMs). The Cold Crucible Induction Melter (CCIM) is an obvious melter choice as it is a demonstrated technology for processing radioactive waste (Russia, France, and South Korea) and can achieve the temperatures required to melt ceramic synroc materials [18,19]. Recently, a first-of-its-kind CCIM test in the United States was conducted at the Idaho National Laboratory (INL) to demonstrate proof-of-principle melt-processing of multiphase ceramic waste forms. The CCIM test was used to 1) produce a ceramic waste form via a scaled but representative process, 2) evaluate the material produced, and 3) assess the feasibility of the CCIM technology. The present work describes recent advances towards demonstrating a viable melt-processing technology for the fabrication of ceramic waste forms from UNF reprocessing.

#### 2. CCIM test summary<sup>1</sup>

In October of 2014, INL personnel melted and attempted to pour a ceramic using a CCIM. Approximately 30 kg of a dry, stoichiometric mixture of oxide and carbonate reagents supplied by the MoSci Corporation was employed as the CCIM feed. Chemical analysis confirmed the composition of the feed material prior to use (Refer to Section 3.1.2). The targeted and measured elemental concentrations are listed in Table 1. For the prepared feed, the measured composition was within typical analytical uncertainty limits of the target composition.

A Ti initiator ring and approximately 13.5 kg of feed was packed into the crucible (10  $\frac{1}{2}$  in. inside diameter) of the CCIM to a height of ~6 in. The Ti initiator ring was suspended off the floor of the

Table 1	
Target and measured elemental concentrations (wt%) in CCIM fee	d material

	Target	Measured
Ва	10.69	10.25
Ca	0.93	0.91
Cd	0.09	0.04
Ce	2.48	2.43
Cs	2.54	2.19
Eu	0.14	0.17
Fe	10.00	9.76
Gd	0.13	0.13
La	1.26	1.17
Мо	0.53	0.37
Nd	4.19	3.74
Pr	1.16	1.18
Rb	0.36	n.m.
Se	0.05	< 0.10
Sm	0.87	0.90
Sn	0.05	0.09
Sr	0.78	1.01
Te	0.49	0.55
Ti	27.55	27.15
Y	0.46	0.44
Zr	2.07	1.98
Sum <sup>a</sup>	66.8	64.5
Batch chemical	Target	Calculated <sup>b</sup>
Batch chemical BaCO3	Target 15.36	Calculated <sup>b</sup> 14.72
Batch chemical BaCO3 CaCO3	Target 15.36 2.31	Calculated <sup>b</sup> 14.72 2.28
Batch chemical BaCO3 CaCO3 CdO	Target 15.36 2.31 0.10	Calculated <sup>b</sup> 14.72 2.28 0.05
Batch chemical BaCO3 CaCO3 CdO CeO2	Target 15.36 2.31 0.10 3.05	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3	Target 15.36 2.31 0.10 3.05 3.11	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3	Target 15.36 2.31 0.10 3.05 3.11 0.16	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3	Target 15.36 2.31 0.10 3.05 3.11 0.16 14.29	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3	Target 15.36 2.31 0.10 3.05 3.11 0.16 14.29 0.15	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3	Target 15.36 2.31 0.10 3.05 3.11 0.16 14.29 0.15 1.48	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3	Target 15.36 2.31 0.10 3.05 3.11 0.16 14.29 0.15 1.48 0.53	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 La2O3 MoO3 Nd2O3 Pr6O11	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m.
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3 SnO2	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3 Sm2O3 SnO2 SrCO3	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07           1.31	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11 1.70
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3 SnO2 SrCO3 TeO2	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07           1.31           0.61	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11 1.70 0.69
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6011 Rb2CO3 SeO2 Sm2O3 SnO2 SrCO3 TeO2 TiO2	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07           1.31           0.61           45.96	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11 1.70 0.69 45.31
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3 SnO2 SrCO3 TeO2 TiO2 Y2O3	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07           1.31           0.61           45.96           0.59	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11 1.70 0.69 45.31 0.56
Batch chemical BaCO3 CaCO3 CdO CeO2 Cs2CO3 Eu2O3 Fe2O3 Gd2O3 La2O3 MoO3 Nd2O3 Nd2O3 Nd2O3 Pr6O11 Rb2CO3 SeO2 Sm2O3 SnO2 SrCO3 TeO2 TiO2 Y2O3 ZrO2	Target           15.36           2.31           0.10           3.05           3.11           0.16           14.29           0.15           1.48           0.53           4.89           1.40           0.49           0.08           1.01           0.07           1.31           0.61           45.96           0.59           2.79	Calculated <sup>b</sup> 14.72 2.28 0.05 2.98 2.69 0.19 13.95 0.15 1.37 0.56 4.37 1.43 n.m. <0.14 1.04 0.11 1.70 0.69 45.31 0.56 2.68

<sup>a</sup> Values are based on conversion of measured elementals to batch components. <sup>b</sup> Measured and calculated values do not include Rb contribution.

<sup>&</sup>lt;sup>1</sup> The experimental parameters and operating conditions during the CCIM test are presented in greater detail elsewhere. (See V. C. Maio, "Production of a Low Temperature SYNROC All Ceramic Surrogate High Level Waste Form in INL's Cold Crucible Induction Melter Pilot –Validation of Test Completion," *U.S. Department of Energy Report INL/MIS-14-34012 (FCRD-SWF-2015-00256)*, Idaho National Laboratory, Idaho Falls, ID (2014).).

crucible with alumina spacers after previous attempts without ring supports resulted in the Ti ring sinking to the bottom of the crucible below the main induction zone before the charge material coupled with the induction frequency. The initial charge was used for primary ignition of the melt. No forced agitation or mixing was used during the test. A melt, as indicated by an optical pyrometer reading of >1700 °C and direct coupling to the material, was achieved in approximately 3 h, at which point additional feed was added at 5.5 kg/hr. The radio frequency (RF) generator of the CCIM was operating at its design power (60 kW) level to maintain the melt and thus, an attempt was made to drain the melter after approximately 40 min (220 min from initiation) and 3 kg of added feed. Nitrogen flow to cool the drain assembly was subsequently stopped and the drain heater was powered on. Less than 5 min after being turned on, the drain heater element shorted out. To counter the loss of heater power, the flow rate of cooling water to the drain assembly was decreased. This immediately led to a water leak in the cooling system, which in turn triggered automatic shutdown of the RF generator and termination of the CCIM test. Fig. 1 shows images of the loading of the CCIM crucible and the melted material.

Cooling water supplied to the crucible stays created a temperature gradient across the melted material during cooling. The estimated cooling rate of material in the center of the crucible was ~6 K/min (averaged over 4 h), and the material near the crucible wall could have realistically cooled at rates as high as 100's K/min. After cooling, seven core samples were taken from the solidified material and characterized.

# 3. Core samples

# 3.1. Experimental

# 3.1.1. Sample preparation

Seven 1-inch diameter core samples were drilled from the solidified material in the CCIM for analysis. Fig. 2 and Table 2 show the core locations and depths, respectively. Because many of the cores cracked during drilling, the cores were re-assembled and held together with thermo-set epoxy prior to sectioning and characterization. A representative image of an as-received, re-assembled core is presented in Fig. 3. Fig. 4 shows scanning electron microscopy (SEM) images of a representative core (sample F) in which dense, columnar, and porous microstructures are evident along the core length. In general, identifying and sectioning samples that were either dense or columnar in appearance was straight-forward. Here, dense core specimens were denoted as *Zone 1*, while columnar samples were labeled *Zone 2*. Accurate identification of



Fig. 2. Digital image showing core sample locations from solidified material in the melter.

Core sample dimensions.		
Core	Core depth (in)	
А	1 <sup>3</sup> /8	
В	1 <sup>1</sup> /2	
С	1 <sup>11</sup> /16	
D	1 3/8	
E	2	
F	1 5/8	
G	1 <sup>1</sup> /4	

the porous sections was more difficult because the transition from a columnar to porous morphology was less definitive. Therefore, samples that could conclusively be identified as porous were labeled as *Zone 3*, and a fourth set of samples consisting of Zone 2 and Zone 3 material, where a definitive distinction was difficult to identify, were labeled *Zone 5*.

#### 3.1.2. Chemical composition

Table 2

Inductively Coupled Plasma-Atomic Emission Spectroscopy (ICP-AES) was used to measure the concentrations of all elements with the exception of Cs, which was measured by Inductively Coupled Plasma-Mass Spectroscopy (ICP-MS). Samples were prepared via sodium peroxide fusion (PF) and lithium metaborate fusion (LM) methods. Both digestion methods were utilized because high concentrations of  $TiO_2$  and  $Cr_2O_3$  are difficult to fully dissolve with the LM fusion method (the preferred digestion



Fig. 1. Images captured inside of CCIM crucible showing (A) the Ti ring on alumina supports, (B) batch material packed under the ring (additional batch material was packed over the ring), and (C) the melted batch material after cooling.



property of the second second

Fig. 3. Core sample F as-received after re-assembly.

protocol for low concentration analysis). Digested samples were analyzed in duplicate for each element of interest by ICP-AES for a total of 4 measurements per element. Cs analysis was measured in duplicate using the same PF digestion solutions used in ICP-AES. The instrumentation was re-calibrated between the duplicate analyses and standards were intermittently measured to ensure the performance of the ICP instruments over the course of the analyses.

# 3.1.3. Fe RedOx

 $Fe^{2+}/Fe^{3+}$  and  $Fe^{2+}/Fe$  (total) ratios were determined by an absorption method using a UV–Vis spectrometer. Samples were first dissolved in a sulfuric-hydrofluoric acid mixture containing ammonium vanadate to preserve the Fe<sup>2+</sup> content. Boric acid was then added to destroy iron-fluoride complexes, and ferrozine was introduced to form ferrous-ferrozine complexes for the determination of Fe<sup>2+</sup> content. Ascorbic acid was then added to reduce Fe<sup>3+</sup> to Fe<sup>2+</sup>, and a second absorbance spectrum was acquired to determine the total Fe content [20]. In addition to UV-Vis-based valence state data, Synchrotron-based hard X-ray microprobe measurements of element distributions were conducted at beam line 2–3 at the Stanford Synchrotron Radiation Lightsource (SSRL). The incident energy was selected using a Si(111) double crystal monochromator, and calibrated by defining the first derivative peak of a reference Fe metal foil to be 7112.0 eV. The current in the storage ring contained 500 mA at 3.0 GeV. The fluorescence lines of interest were measured using a Vortex Si-drift detector (Hitachi) using Xspress3 pulse processing electronics (Quantum Detectors). The incident x-ray beam was focused to a size of  $2 \times 2$  microns using Rh coated Kirkpatrick-Baez mirrors (Xradia Inc.). The x-ray fluorescence (XRF) intensity was mapped by rastering the sample in the focused beam. The XRF maps were performed at several energies across the Fe K-edge to preferentially excite different Fe species and other elemental species. Internal variation within the sample was used to differentiate the composition of the sample, rather than referencing to specific crystalline standards. This process was sufficient to identify the oxidation state and along with the remaining elemental compositions, make inferences on total composition.

## 3.1.4. Phase identification and microstructure

Samples were characterized via x-ray diffraction (XRD, D8 Advance, Bruker AXS Inc., Madison, WI) to identify the resulting phase(s). Portions of each sample were initially ground in an automatic Spex mill for 4 min using an agate cup and media. The powders were subsequently hand ground with an agate mortar and pestle in alcohol and mounted to a glass slide using a collodion/ amyl acetate solution. The XRD patterns were collected from 5 to 70° 2 $\theta$  with a 0.02° step size and scan rate of 1 s/step. XRD patterns for refinement were collected from 10 to 70° 2 $\theta$  at a scan rate of 1°/ min (Rigaku D/Max 2100 Powder X-ray Diffractometer), and analyzed using the Jade Software package to estimate phase abundances.

SEM and Energy Dispersive Spectroscopy (EDS) measurements were performed with a Hitachi SU6600 instrument at the Advanced Materials Research Lab of Clemson University. EDS maps were collected for Cs, Ba, Mo, Zr, Cr, Al, Fe, Nd, Ca, La, Ce, and Ti. Complementary imaging and EDS analysis were carried out at SRNL with a Hitachi TM3000 SEM.

Transmission electron microscopy (TEM)/scanning transmission electron microscopy (STEM)/Energy-dispersive X-ray spectroscopy (EDX) was performed using a FEI Tecnai F30 electron microscope operated at 300 kV at Los Alamos National Laboratory. Samples were first mechanically polished to a thickness of less than 10  $\mu$ m, at which point a precision ion polishing system (PIPS) was employed to yield electron transparent (less than 200 nm in thickness) specimens. EDX spectra were used to identify the chemical composition of observed crystalline phases, while selected area electron diffraction (SAED) patterns and high resolution TEM observations were utilized to determine the crystalline nature and structure of the phases.

#### 3.1.5. Chemical durability

A crushed sample leach test (Product Consistency Test (PCT) Method-A) was performed in triplicate to assess aqueous chemical durability.<sup>2</sup> Also included in the experimental test matrix were Environmental Assessment (EA) benchmark glass [21], the Approved Reference Material (ARM), the Low-activity Reference Material (LRM), and blanks from the sample cleaning batch. All standards were ground, washed, and prepared according to the standard PCT procedure [22]. Fifteen milliliters of water were added to 1.5 g of sample or standard in stainless steel vessels. The vessels were closed, sealed, and placed in an oven at  $90 \pm 2$  °C; the samples were maintained at this temperature for 7 days. Once cooled, the resulting solutions were sampled (filtered and acidified), and analyzed. Samples of a multi-element, standard solution were also included as a check on the accuracy of the ICP-AES instrument used for these measurements. Based on the measured compositions, fractional elemental release from the samples was calculated using the equation

<sup>&</sup>lt;sup>2</sup> The PCT has been used in melt-processed ceramic development research as a convenient method to measure specific elemental releases and identify phases with low durability. At this time, the test is not being used to provide quantitative durability values.



Fig. 4. SEM phase contrast images that reveal microstructure differences along the length of the cores.

$$FR = \frac{C_i \times V_s}{m_s \times f_i} \tag{1}$$

where  $C_i$  is the concentration of element "*i*" (e.g., Cs) in solution (g/ L),  $V_s$  is the volume of leachant solution (L),  $m_s$  is the mass (g) of sample, and  $f_i$  is the fraction of element "*i*" in the unleached sample <sup>3</sup>(unitless). From Eq. [1], the fractional release (*FR*) of element "*i*" is obtained as a figure of merit for the release of each element. Such a methodology was used to facilitate a comparison of the relative release among the samples.

#### 3.2. Results & discussion

#### 3.2.1. Chemical composition

Sample from each core was selected for chemical analysis and grouped according to microstructure (see Section 3.1.1 for morphology categorization details). For a suitable representation of the bulk material, one additional sample was measured that was not sectioned prior to analysis. In total, 29 samples taken from 4 different cores and from various (~7) microstructure regions (i.e. zones) were analyzed.

Table 3 lists the measured elemental concentrations in the bulk CCIM material, i.e., the sample not sectioned and analyzed by zone. With the exception of Al, Cs, Mo, and Te, the results indicate that the CCIM material composition was, in general, comparable to the target composition. Although no Al was added to the CCIM feed material, Al<sub>2</sub>O<sub>3</sub> is known to react with the composition. Consequently, it is suspected that the Al<sub>2</sub>O<sub>3</sub> supports used to hold the Ti initiator ring in place <sup>4</sup> reacted with the ceramic mixture during CCIM processing, thereby contributing to a measured Al concentration in the melted material. Cs is a highly volatile species and previous experiments have generally shown that about 50% of

#### Table 3

Measured elemental concentrations (wt%) in the bulk CCIM material not separated	d
by a zone.	

Element	Target	Measured
Al	0.00	2.21
Ba	10.69	10.54
Ca	0.93	0.93
Cd	0.09	<0.10
Ce	2.48	2.47
Cs	2.54	1.73
Eu	0.14	0.12
Fe	10.00	9.88
Gd	0.13	0.12
La	1.26	1.22
Мо	0.53	0.15
Nd	4.19	4.02
Pr	1.16	1.21
Rb	0.36	n.m.
Se	0.05	<0.10
Zr	2.07	1.97
Sm	0.87	0.85
Sn	0.05	n.m.
Sr	0.78	0.99
Те	0.49	0.13
Ti	27.55	30.13
Y	0.46	0.40

the Cs is lost during heating. However, it appears that less Cs, about 32%, was lost in this CCIM test. The difference in measured Cs retention may be accounted for by a cold cap (the CCIM was fed during production) and the reduced surface area to volume ratio in the CCIM compared to lab testing. Substantial amounts (>70%) of Mo and Te were unaccounted for in the measured analysis.

Table 1 indicates the feed material could have been deficient in Cs and Mo by as much as ~15% and ~30%, respectively. Therefore, the Cs and Mo retention would be greater than if calculated from Table 3. Nevertheless, significant evaporation of Mo, Te, and Cs, is plausible considering the melting temperatures of MoO<sub>3</sub>, TeO<sub>2</sub>, and Cs<sub>2</sub>CO<sub>3</sub> are all in the range of 700–800 °C and is ~500 °C for Cs<sub>2</sub>O [23]. Furthermore, significant sublimation of pure MoO<sub>3</sub> and TeO<sub>2</sub> above the melting temperature has been reported [24,25]. The physical characteristics (i.e. low melting point and high vapor pressures at elevated temperatures) likely account for the lower than expected concentrations of Te, Mo, and Cs in the CCIM material.

Chemical composition was also measured as a function of vertical (zone) and radial location in the melter. The measured elemental concentration for each zone is plotted in Fig. 5. In general, chemical composition did not depend on the zone from which the material came, indicating that the CCIM material was relatively homogeneous. However, the data presented in Fig. 5 show that Al and several lanthanide series elements exhibited a greater variation in their means compared to other elements. Analysis of variance (ANOVA) was used to test for statistically significant differences in the means of the elemental concentrations between zones. ANOVA identified significant differences (at the 5% significance level) for Ce, La, Nd, Pr, Sr, and Te<sup>5</sup>. Specifically, the means for Ce, La, and Pr were not significantly different between zones 1, 2, and 5, or between zones 3 and 5 (i.e. Zone 5 was common). Similarly, the means for Nd and Sr were not significantly different between zones 1, 2, and 5 or between zones 2, 3, and 5 (i.e. zone 2 and 5 were common). The ANOVA results indicated that lanthanide series elements concentrations were dependent on vertical location (zone). Indeed, when individual element mean concentrations are plotted as a function of zone, as shown in Fig. 6, La, Ce, Pr, Nd, Eu, Y, and Sr concentrations are observed to increase with vertical height in the cores. The trend in the lanthanide series elements is accompanied by a decrease in concentration of Al with vertical height in the cores but, all the other constituent elements do not exhibit any recognizable trend as a function of zone. The variance in the Al concentration is unknown. However, the closed-end Al<sub>2</sub>O<sub>3</sub> cylinders used to support the Ti initiator ring during processing were observed to significantly react with the feed material, as evidenced by considerable dissolution and dimensional changes in one of the Al<sub>2</sub>O<sub>3</sub> supports found during core drilling. Although Al is typically a 3 + cation, it's ionic size is relatively small and direct substitution for the larger 3+ rare-earth cations would not be expected. However, one possible explanation can be considered if the bottom of the melt cooled first and initiated a crystallization front proceeding upwards and following the crystallization sequence reported previously in which hollandite crystallization precedes the others upon cooling [26]. This could in turn lead to a slightly higher percent of hollandite at the bottom of the CCIM (in agreement with Fig. 9) and thus, a lower rare-earth element concentration in that region. Alternatively, it is possible that Al may form a phase(s) or induce

<sup>&</sup>lt;sup>3</sup> In all cases, measured elemental concentrations were used as opposed to target concentrations.

<sup>&</sup>lt;sup>4</sup> Refer to reference in footnote a for a discussion of the initiator ring setup.

<sup>&</sup>lt;sup>5</sup> The ANOVA for Te indicated significant differences between zone 2 and 5 but, zone 1 and 3 were common. This result does not appear to support the composition trend in the core samples and is ultimately unknown. Analysis of additional and a larger data set would be needed to confirm and explain the result.



Fig. 5. Plot of mean elemental concentrations and corresponding standard deviations from samples grouped according to morphology (Zone) type.

reactions making the rare-earth containing phases less energetically favorable. In contrast to the ANOVA results for Ce, La, Nd, Pr, Sr, a linear correlation between the Te composition and location in the core was not observed. The differences found in the Te means could be correlated to an unknown phenomenon.

In practical terms, the ANOVA results, the subjectivity involved in differentiating distinct boundaries between zones (morphology transitions, in particular between zones 3 and 2, hence the use of zone 5), and the microscopy results (Section 3.2.2) indicate most rare-earth elements are likely distributed less uniformly than other elements. Nevertheless, the magnitude of those differences remains small; the variances in the elemental means shown in Figs. 5 and 6 are typically less than 0.5 wt%.

Fig. 7, Fig. 6 shows the measured mean elemental concentrations for different core samples. In contrast to Fig. 5, the data in Fig. 7, Fig. 6 are grouped according to radial position in the melter. In general, smaller standard errors (relative to the zone analysis) were observed across the cores with the exception of core F, which exhibited the greatest variation. The variation in core F was greatest for many of the lanthanide series elements and may be related to zone 3 elemental partitioning. An additional statistical analysis revealed that the ANOVA results for zone grouping were not affected by the Core F zone 3 measurements, i.e., nothing that would bias previous conclusions was found. Given that Core F was located near the wall of the crucible, it is possible that cooling gradients and interactions with the cold wall of the melter influenced elemental partitioning. Overall, the chemical composition measurements were remarkably consistent. The results confirm that the CCIM produced a melt that was chemically homogenous both radially and vertically. Based on these results, canister to canister homogeneity would be expected to be good but would need to be confirmed over extended operations.

## 3.2.2. Fe RedOx

Reducing conditions are known to have positive effects on the final properties of Synroc-type ceramic materials [27,28]. Ti metal was added to the CCIM during initiation, and it was expected that some amount of reduction was possible. Fe redox was measured on one sample, Core D, to determine the extent of reduction in the CCIM material. As with the other analyses, Fe redox was measured for different zones (microstructures) within the core. The results, summarized in Table 4, reveal that the CCIM material was partially reduced (~20% Fe<sup>2+</sup>).

The distribution of Fe species in ceramic waste forms has not yet



Fig. 6. Plots of elemental mean concentrations as a function of zone.

been extensively studied. In this work, mapping of the Fe fluorescence was performed as five discrete energies through the Fe Kedge (7113, 7120, 7122, 7130, and 7133 eV) which were believed to be most sensitive to the Fe oxidation states expected. These maps were combined and principal component analysis (PCA) was performed to determine the spatial regions that showed the greatest chemical variation. Several X-ray absorption near edge spectra (XANES) were measured to confirm the chemical variation of Fe. From this analysis, three major components, two oxidized Fe and one reduced Fe, were found, with oxidation state assignments based on edge position as shown in Fig. 8 (spots 1 5 and 7). The intensities of the spectra were used as endmembers to perform linear combination fitting at each point of the multi-energy maps. The sum of the two oxidized species was used as the  $Fe^{3+}$  concentration in the sample. Procedures regarding PCA identification from image data [29,30] and least squares fitting of the XANES [31] have been detailed elsewhere. The results from fitting the spectra that are shown in Fig. 9 indicate that Fe<sup>3+</sup> partitions primarily with elements found in the hollandite phase (i.e. Ba, Ti, Cs, and Al), while Fe<sup>2+</sup> partitions primarily with the Nd-containing phase. Further analysis into the nature of the Fe<sup>2+</sup>-containing phase(s) is ongoing. However, the ionic radius of an eight-fold coordinated  $Fe^{2+}$  is likely too small to substitute for  $Ca^{2+}$  in zirconolite [32] and  $Fe^{2+}$  is not expected to readily substitute in simple perovskite structure (ABO<sub>3</sub>) based on ionic radii considerations. However, Fe<sup>2+</sup> may substitute on the A site in pyrochlore and as such, a complex pyrochlore phase with both  $Nd^{3+}$  and  $Fe^{2+}$  may be forming in addition to other reduced Fe titanate compounds.

#### 3.2.3. Phase identification and microstructure

XRD confirmed three primary phases in the core samples as having hollandite, pyrochlore/zirconolite, and perovskite structures. Phase analysis conducted on the different morphologies observed in the individual cores indicated similar phase assemblage. In general, the XRD results indicated material in the CCIM was homogenous at varying depths in the melt as well as radially from the center to the edge of the melt. Phase abundances calculated from Rietveld refinement of the XRD patterns are summarized in Fig. 10. In general, the relative phase abundances indicated the CCIM test was successful in producing a homogenous ceramic from a melt.

SEM backscattered electron (BSE) images of Zone 1 (dense), 2 (columnar), and 3 (porous) areas from Core D are shown in Fig. 11 along with corresponding EDS elemental maps for zone 2. This set of BSE images and elemental maps are representative of all the core samples. In the BSE images, 3 to 5 distinct contrast phases are visually evident and no significant differences in elemental partitioning are evident in the maps, as expected, based on chemical composition and XRD analyses. Fig. 11 also summarizes an averaged semi-quantitative elemental analysis that was performed for the various phases based on phases with similar chemical makeup but from several samples. Phases were identified based on the elements that measured greater than 1 atom %. Notably, the hollandite phase contains Al, which is known to readily substitute into hollandite structures and is used to stabilize pure Fe-hollandite ceramics [12,13,27]. Hollandite is most distinctly identified in the Ba and Ti maps. Pyrochlore, zirconolite, and perovkite phases identified in XRD were less straightforward to identify using the SEM and EDS information. Specifically, Ca and Zr often appeared to coexist with lanthanide series elements in multiple phases making it impossible to distinguish zirconolite (CaZrTi<sub>2</sub>O<sub>7</sub>), pyrochlore  $(A^{3+}_{2}Ti_{2}O_{7})$ , and perovskite  $(A^{2+}TiO_{3})$  based on the composition data alone. These phases are most distinctly identified in the Ca, Zr, and Nd maps but it is also evident that significant elemental substitution exists in the samples making interpretation more



Fig. 7. Plot of mean elemental concentrations and corresponding standard deviations from samples grouped according to radial distance. (D: center, E: ½ radius, F: edge, B: ½ radius).

 Table 4

 Measured Fe<sup>2+</sup> fraction in core D.

	Fe <sup>2+</sup> /Fe total
Zone 1	0.13
Zone 2	0.19
Zone 5	0.20
Zone 4	0.24

complex.

The SEM/EDS data also revealed Fe-rich and Cs-rich phases in addition to the major phases. The Fe-rich phases appeared to contain Al and possibly minor amounts of Ba. This result was not unexpected considering that various parasitic phases containing Fe, Al, or Ba have been observed in previous work [27]. Furthermore, because the ceramic in this test was formulated for a pure Fe-hollandite, excess Fe would be available from its replacement by Al (introduced during processing) in the hollandite phase to from parasitic compounds (and substitute in pyrochlore/zirconolite phases). Several Cs-rich phases have also been observed in previous work. In particular, non-durable Cs-Mo containing compounds and Cs-Al titanates are known to form. Interestingly, Cs-Mo containing phases did not appear the dominant Cs-rich phase in the core samples. Instead, Cs appeared in combination with other elements, most markedly Al and possibly various lanthanides. As noted previously, the Fe-hollandite formulation does not represent an optimal waste form composition but was selected for this initial proof-of-principal testing due to its lower melting temperature behavior and compatibility with the current INL CCIM design.

TEM/STEM was performed on Core F Zone 1 to further investigate the chemistry and crystallinity of phases in the CCIM material. Fig. 12 shows three sets TEM/STEM images, EDX spectra, and SAED patterns. Each TEM image was taken across multiple phases in order to investigate the boundary region and the beam was aligned with a different phase in each image. In all images, the phase boundary is relatively sharp with no clear secondary phases at the boundaries. In Fig. 12 (A) the beam was aligned with the phase identified as hollandite based on chemical composition obtained from EDX and crystalline symmetry, identified as tetragonal, from the SAED pattern. In Fig. 12 (B) the beam was aligned with the phase



Fig. 8. XANES data (A) collected at different spots (B) with different Fe intensities. High Fe concentrations areas are indicated by red color. (For interpretation of the references to colour in this figure legend, the reader is referred to the web version of this article.)



**Fig. 9.** 2-D XANES plots showing Ti, Nd, Ba, Cs, Al, Ca, and Fe elemental distribution in Core D Zone 2. Distribution of  $Fe^{2+}$  and  $Fe^{3+}$  is also shown. The bi-color Fe plot is a composite (not total) of the  $Fe^{2+}$  (green) and  $Fe^{3+}$  (red) plots. The intensity scale bar corresponds to all plots but the absolute counts varied depending on element. (For interpretation of the references to colour in this figure legend, the reader is referred to the web version of this article.)

# 500µm



Fig. 10. Summary of primary phase abundances in core samples compared to estimated abundances based on feed composition and target phase assemblage.

identified as perovskite based on chemical composition obtained from EDX and crystalline symmetry from SAED pattern. This phase is possibly a (Nd,Sr,Ca)(Ti,Fe)O<sub>3</sub> solid solution with the perovskite structure, which possess a monoclinic cell in a three-tilt system. In Fig. 12 (C) the beam was aligned with the phase identified as zirconolite based on chemical composition obtained from EDX and crystalline symmetry from SAED pattern. The (Ca,Nd)Zr(Ti, Fe)<sub>2</sub>O<sub>7</sub> zirconolite phase was pseudo-cubic and is similar to pyrochlore which is an anion deficient superstructure based on the fluorite structure (MX<sub>2</sub>), and occurs as various polytypes including monoclinic and trigonal. The major differences between zirconolite and pyrochlore, are the coordination number of Zr and the pyrochlore is expanded (by ~ 4%) along the zirconolite C axis (the pyrochlore [001] axis) relative to zirconolite. In another words, pyrochlore can be thought of as an expanded zirconolite.

# 3.2.4. Chemical durability

Chemical durability was performed to assess the relative durability of different areas from the CCIM material. Specifically, potential differences in durability throughout the bulk material were tested by sampling material from different Zones (similar to the chemical composition testing). The results of the aqueous durability testing are summarized in Fig. 13. Fractional release (FR) of elements not shown in Fig. 13 were below detection (<0.1 ppm) in the leachate. Similar to the composition data, the greatest standard error was observed in the Zone 3 and 5 samples suggesting that precise sampling in those areas is difficult or that those areas possess a greater degree of inhomogeneity. The FR results indicate Cs and Al elemental releases were practically constant across all zones. The Te FR exhibited comparatively more variation from each zone. In particular, Zone 1 (dense) exhibited a higher Te FR compared to the other zones. Mo FR exhibited the greatest variation than the other elements in each zone. Recalling the SEM/EDS results, Mo appeared distributed throughout the ceramic, but may not have been chemically or uniformly incorporated into the primary phases. It is known that water soluble Cs, Ca, and Sr molybdates can form in synrock-type materials and these compounds may be a possible explanation for the Cs and Mo release rates. However, the lack of correlation between Cs and Mo *FR* and the low release of Sr and Ca indicate further explanation is needed.

# 4. Conclusions

The first scaled proof of principle test for melt-processing multiphase crystalline waste forms for a combined UNF waste stream (Cs/Sr, fission products) was completed using a CCIM. Characterization of material fabricated during the test indicated that the ceramic material exhibited a desirable phase assemblage consisting primarily of hollandite, zirconolite/pyrochlore, and perovskite phases. The hollandite phase exhibited relatively narrow substitution of other elements as evidenced by the XANES, EDS/EDX data. In particular, the primary cations (excluding Ti which is in all phases) Ba, Cs, and Fe (3+) were well confined to the hollandite phase as evidenced in the XANES and EDS spectra. Comparatively, pure zirconolite and perovskite (A<sup>+3</sup>) were not found and are not believed to exist in any appreciable amount. The close correspondence among Ca, Zr, and Nd in the data suggest complex phase reactions among the zirconolite, pyrochlore, and perovskite phases. In the zirconolite structure,  $Ca^{2+}$  or  $Zr^{4+}$  can be substituted by Nd<sup>3+</sup>, with the former being preferred due to the smaller disparity in ionic radii for  $Ca^{2+}$  and  $Nd^{3+}$  [33]. In the case of  $Ca^{2+}$  substitution by Nd<sup>3+</sup>, charge compensation can occur by either trivalent cation substitution at  $Zr^{4+}/Ti^{4+}$  sites, or via metal vacancy formation at Ca<sup>2+</sup>/Zr<sup>4+</sup> sites [34]. Furthermore, extensive Nd substitution into the zirconolite structure leads to partitioning into pyrochlore and/ or perovskite phases. While more precise determinations of compound stoichiometry in the Ca-Nd-Zr areas of the melt-processed samples are on-going, it is possible that many of these regions have nominal compositions of Ca<sub>1-x</sub>Zr<sub>1-x</sub>Nd<sub>2x</sub>Ti<sub>2</sub>O<sub>7</sub>, where partitioning into various doped perovskite and pyrochlore phases occurs at critical Nd concentrations.



- (Al,Fe)<sub>1.70</sub>Ti<sub>1.28</sub>O<sub>5.0</sub> (FeAlTiO<sub>5</sub>) Spot 2
- Spot 3  $Ba_{0.31}Cs_{0.19}Fe_{1.2}Al_{0.76}Ti_{6.4}O_{16}$  (hollandite) Spot 4

(Zr,Fe,Nd,Ca,Al,Ce)<sub>1.7</sub>Ti<sub>2.0</sub>O<sub>7.0</sub> (pyrochlore/zirconolite)

Fig. 11.  $600 \times$  magnification EDS elemental maps for Core F Zone 2.





Fig. 12. TEM images of phase boundaries with the beam aligned to (A) hollandite, (B) perovskite, and (C) zirconolite/pyrochlore phases. SAED patterns and EDX spectra correspond to the phase aligned with the beam (and in "white" text).



Fig. 13. Fractional release for elements with measureable response after exposure to PCT. Individual samples are indicated with "x" and corresponding standard error bars are drawn.

7-day tests performed on multiple samples from the fabricated material resulted in measureable release for Al, Cs, Mo, and Te species. The fractional release of these elements were on the order of 0.001–0.1 of the starting concentration. Overall, the variability in the fractional release indicates chemical and microstructural homogeneity.

This CCIM test improves our understanding and the feasibility of using an induction melter to process a ceramic waste form from a melt. Specifically, canister to canister homogeneity (production consistency) would appear to be excellent and although complex, the crystallization of the phases and the resulting elemental partitioning produced an acceptably durable ceramic waste form that incorporated constituents from a combined UNF waste stream. This test substantiates the importance and practicality of using induction melters to fabricate crystalline ceramic waste forms.

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