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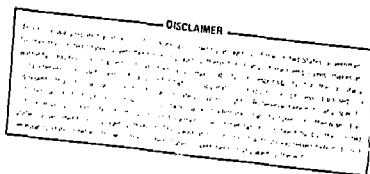
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Investigations on SYNROC Mineralogy

Progress Report to the Lawrence Livermore  
National Laboratory, As specified in Contract  
No. 2248301


A. E. Ringwood  
Research School of Earth Sciences  
Australian National University



January 21, 1982

  
Lawrence  
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## FACTORS INFLUENCING THE LEACHING PERFORMANCE OF HOLLANDITE

A.E. Ringwood, S.E. Kesson, V.M. Oversby and W. Hibberson

### 1. Introduction

An earlier report by A.N.U. (Ringwood et al., 1981a) studied the leaching of cesium from hollandite and SYNROC-C formulations. It was found that cesium was readily leached from samples hot-pressed under oxidizing conditions (near Ni-NiO) because cesium was not successfully immobilized in the hollandite lattice but had combined with molybdenum from the waste forming soluble cesium molybdate. This problem no longer occurred if hot-pressing was carried out under more reducing conditions (between Co-CoO and Fe-FeO) because molybdenum was reduced from the hexavalent state to  $\text{Mo}^{4+}$  and/or to a metal alloy; and  $\text{Cs}_2\text{MoO}_4$  was destabilized.

However in subsequent experiments with hollandite and SYNROC formulations which contained no molybdenum, the leaching behaviour of cesium was still found to be very erratic, varying by as much as an order of magnitude, depending on hot-pressing redox conditions and sample preparation methods. This report evaluates the effects of such physical and chemical parameters on hollandite leaching performance and attempts to explain this in terms of hollandite crystal chemistry and reaction kinetics.

### 2. Starting materials and experimental methods

Table 1 lists the compositions of the various hollandite preparations used in this study. These were prepared by a range of techniques, and by three independent laboratories (see Table 1). Run products were examined by microprobe and x-ray techniques, and leaching tests carried out as described by Oversby and Ringwood (1981).

In most of the experimental studies described herein, microprobe investigations were largely directed at the one composition, "Cs-holl.mix 1".

This composition (Table 1) corresponds to  $\text{BaAl}_2\text{Ti}_6\text{O}_{16}$  (70 wt%) +  $\text{TiO}_2$  (30 wt %) + an additional 2%  $\text{Cs}_2\text{O}$ . The excess of  $\text{TiO}_2$  is present in order to maximise solid solution in the hollandite towards the hypothetical Ti-end-member " $\text{Ba}_6\text{Al}_2\text{Ti}_8\text{O}_{16}$ ". Similar solid solutions occur in hollandite in SYNROC, because of the high  $\text{TiO}_2$ -activity of the SYNROC phase assemblage.

### 3. Effect of hot-pressing temperature on leach-rate

The composition "Cs-holl.mix I" was dried in air at  $800^\circ\text{C}$ . An additional 2% Ti metal was incorporated immediately prior to hot-pressing, in order to facilitate redox control (this topic is discussed further in Section 4 (b)). The samples were hot-pressed in Ni capsules at temperatures ranging from 1100 to  $1350^\circ\text{C}$ . The leach-rates obtained for these samples (Table 2) are illustrated in Fig.1. As in our earlier studies (Ringwood et al., 1981b) the leach-rates are observed to decrease markedly with time, falling by about an order of magnitude over seven days. The individual samples themselves yield leach-rate curves dispersed over about an order of magnitude (Fig.1). However it is apparent that the highest leach-rates occur in samples hot-pressed at temperatures of  $1300^\circ\text{C}$  or higher. The less-satisfactory leaching performance of these latter samples is tentatively attributed to incipient partial melting at near-solidus temperatures. Under such conditions some cesium may partition into an intergranular melt phase, from which it is then more readily leached.

### 4. Effects of hot-pressing redox conditions on leach-rates

#### (a) Redox conditions controlled by sample capsule

The composition "Cs-holl.mix I" (Table 1) was hot-pressed in Ni capsules at 5 kbar and temperatures from  $1150^\circ\text{C}$  to  $1275^\circ\text{C}$ . The same starting material was then hot-pressed in graphite dies at 4000 psi and

Table 1. Bulk compositions of hollandite starting materials

	S24 and S24-2A	S25 and S25-2A	S16	S03	H1-4	"Cs-holl. mix I"	"Cs-holl. mix II"
TiO <sub>2</sub>	73.6	72.3	63.9		71.1	74.2	67.8
Al <sub>2</sub> O <sub>3</sub>	13.2	13.0	13.7		11.0	9.5	17.9
BaO	11.1	9.2	18.6		14.9	14.3	12.3
Cs <sub>2</sub> O	2.3	5.6	3.8		3.0	2.0	2.0
Sum	100.	100.	100.		100.	100.	100.

Notes:

S24 corresponds to a composition with molar  $(\text{BaO} + \text{Cs}_2\text{O}) : (\text{Al}_2\text{O}_3) : (\text{TiO}_2) = 1 : 1.6 : 11.4$  and molar  $\text{Cs}_2\text{O} : \text{BaO} = 1:9$ . Prepared by L.L.L. from spray-dried nitrates (Memo; 2 February 1981).

S25 corresponds to a composition with the above molar ratios except  $\text{Cs}_2\text{O} : \text{BaO} = 1:3$ . Prepared by L.L.L. from spray-dried nitrates (Memo; 2 Feb. 1981).

S16 prepared by L.L.L. from spray-dried nitrates (Memo; 2 Feb. 1981). S16 is "stoichiometric" with molar  $(\text{BaO} + \text{Cs}_2\text{O}) : (\text{Al}_2\text{O}_3) : \text{TiO}_2 = 1:1:6$ .

S03 prepared by L.L.L. from spray-dried nitrates (Memo; 2 Feb. 1981) with molar  $(\text{BaO} + \text{Cs}_2\text{O}) : (\text{Al}_2\text{O}_3) : \text{TiO}_2 =$

S24-2A and S25-2A are Sandia preparations of compositions S24 and S25 (letter; 27 Jan. 1981).

H1, H2, H3 and H4 correspond to four different preparations of the S16 hollandite composition to which an additional 20% of TiO<sub>2</sub> has been added. Their bulk composition is as shown. Preparations by R. Dosch, Sandia Labs., (Memo; 2 Dec. 1980).

"Cs-holl. mix I" corresponds to: 70%  $\text{BaAl}_2\text{Ti}_6\text{O}_{16}$  + 30% TiO<sub>2</sub> + additional 2% Cs<sub>2</sub>O (by weight).

"Cs-holl. mix II" corresponds to: 60%  $\text{BaAl}_2\text{Ti}_6\text{O}_{16}$  + 30% TiO<sub>2</sub> + 10% Al<sub>2</sub>O<sub>3</sub> + additional 2% Cs<sub>2</sub>O.

Fig.1: Effect of hot-pressing temperature  
on leach-rate

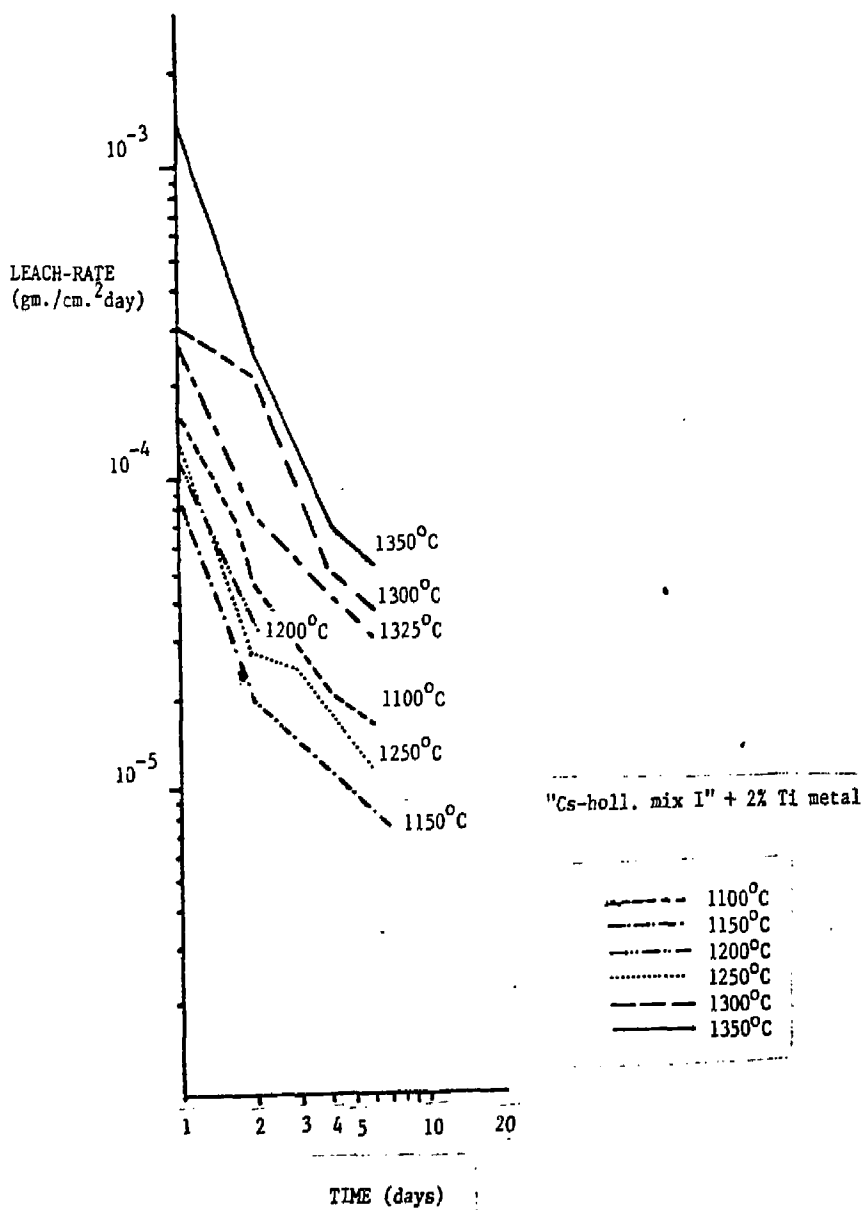


Table 2. Effect of hot-pressing temperature on cesium leach-rate

"Cs-holl. mix I" + 2% Ti metal hot-pressed in Ni capsules for 1 hour at temperatures shown

<u>T°C</u>	<u>Day 1</u>	<u>Day 2</u>	<u>Day 3</u>	<u>Day 4</u>	<u>Day 6</u>
1100	$1.7 \times 10^{-4}$	$4.6 \times 10^{-5}$		$2.0 \times 10^{-5}$	$1.6 \times 10^{-5}$
1150	$8.4 \times 10^{-5}$	$1.9 \times 10^{-5}$	$2.2 \times 10^{-5}$	$1.1 \times 10^{-5}$	$7.4 \times 10^{-6}$
1200	$1.2 \times 10^{-4}$	$3.3 \times 10^{-5}$			
1250	$1.4 \times 10^{-4}$	$2.7 \times 10^{-5}$	$2.4 \times 10^{-5}$	$1.9 \times 10^{-5}$	$1.1 \times 10^{-5}$
1300	$3.1 \times 10^{-4}$	$1.1 \times 10^{-4}$		$5.0 \times 10^{-5}$	$3.8 \times 10^{-5}$
1325	$2.8 \times 10^{-4}$	$7.6 \times 10^{-5}$		$4.2 \times 10^{-5}$	$3.2 \times 10^{-5}$
1350	$1.4 \times 10^{-3}$	$2.6 \times 10^{-4}$		$6.9 \times 10^{-5}$	$5.0 \times 10^{-5}$

Table 3. Cesium leach-rates for hollandite preparations hot-pressed in graphite or nickel

Experimental conditions: Graphite die, 4000 psi, 1 hour, temperatures as shown

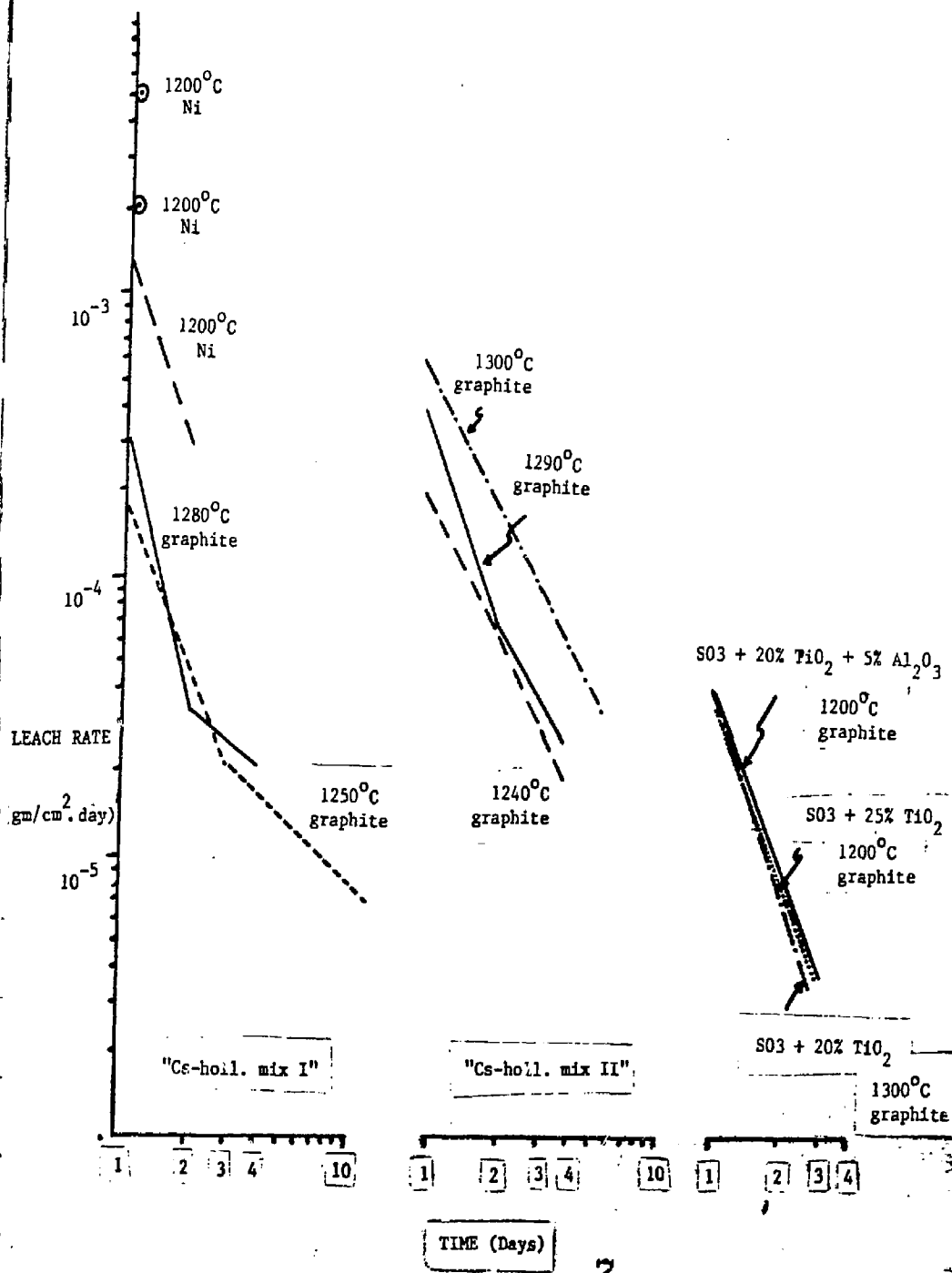
<u>Hollandite preparation</u>	<u>T°C</u>	<u>Day 1</u>	<u>Day 2</u>	<u>Day 3</u>	<u>Day 4</u>	<u>Day 6</u>	<u>Day 11</u>
"Cs-holl. mix I"	1250	$1.8 \times 10^{-4}$				$1.3 \times 10^{-5}$	$6.9 \times 10^{-6}$
"	1280	$3.1 \times 10^{-4}$	$3.4 \times 10^{-5}$		$2.1 \times 10^{-5}$		
"Cs-holl. mix II"	1240	$2.1 \times 10^{-4}$	$7.4 \times 10^{-5}$		$1.8 \times 10^{-5}$		
"	1290	$4.3 \times 10^{-4}$	$7.8 \times 10^{-5}$		$2.4 \times 10^{-5}$		
"	1300	$6.2 \times 10^{-4}$		$1.0 \times 10^{-4}$		$3.1 \times 10^{-5}$	

Experimental conditions: Nickel capsule, 5 kbar, 1 hour

"Cs-holl. mix I"	1200	$1.4 \times 10^{-3}$	$2.7 \times 10^{-4}$
"	"	$1.9 \times 10^{-3}$	
"		$5.4 \times 10^{-3}$	



Figure 2. Comparative leaching performance of hollandite preparations hot-pressed in graphite or nickel.



temperatures of 1250 and 1280°C. The products of these two series of experiments were leached at 95°C in H<sub>2</sub>O, and subjected to microprobe analysis.

The leaching results are presented in Fig.2 and Table 3. The products of hot-pressing in Ni capsules display high initial leach rates, typical of hollandite hot-pressed under these conditions (compare with Fig.3). They lost a substantial proportion of their total complement of cesium on the first day, and for two of the three samples, no further leaching was carried out. In contrast, the leaching performance of the two samples hot-pressed in graphite is markedly improved, even the first day's leach rate being about an order of magnitude less than that for the samples hot-pressed in Ni.

Microprobe analyses of the runs (Tables 4 and 5) reveal that the product hollandites typically exhibit a range in composition, as indicated in the tables. In particular, cesium varies from below detection limits to an ideal maximum value of ~2%. (Mass-balance considerations suggest that cesium is not fully immobilized in hollandite; minor amounts of an unidentified Cs-rich phase may also be present in runs in both Ni and graphite.)

The most striking difference between the two sets of product hollandites is apparent in the unusually low Al<sub>2</sub>O<sub>3</sub> contents and "non-stoichiometric" structural formulae of the samples hot-pressed in graphite. (Compare Tables 4 and 5.) Moreover, significant quantities of an aluminous phase (corundum or Al<sub>2</sub>TiO<sub>5</sub>) as well as rutile, are observed in the latter series.

In order to evaluate the structural formulae, it is important to bear in mind that there are three ideal end-members contributing to the hollandite solid solution series, namely:

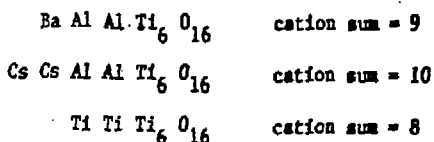


Table 4. Hot-pressing of "Cs-holl. mix I" composition in Ni capsules.

T <sup>o</sup> C	1275 <sup>o</sup>		1250 <sup>o</sup>		1200 <sup>o</sup>		1150 <sup>o</sup>	
	<u>Hollandite</u>		<u>Hollandites.</u>		<u>Hollandites</u>		<u>Hollandites</u>	
TiO <sub>2</sub>	65.1		62.8 66.1		67.8 70.2		62.7 66.6	
Al <sub>2</sub> O <sub>3</sub>	12.4		14.1 14.3		14.1 14.8		14.5 13.8	
BaO	20.6		21.5 19.9		16.1 15.3		20.4 20.5	
Cs <sub>2</sub> O	<u>2.1</u>		<u>1.6</u> <u>0.2</u>		<u>1.8</u> <u>.2</u>		<u>2.1</u> <u>.2</u>	
Sum	100.2		100.0 100.5		99.8 100.3		100.2 100.9	
coexisting phases	rutile rare corundum		rutile		rutile		rutile	

Structural formulae based on 16 oxygens

Ti	6.103	5.897	6.001	6.107	6.132	5.877	6.043
Al	1.822	2.075	2.035	1.990	2.026	2.123	1.963
Ba	1.006	1.052	0.941	0.756	0.696	0.996	0.969
Cs	<u>0.112</u>	<u>0.085</u>	<u>0.010</u>	<u>0.092</u>	<u>0.0</u>	<u>0.112</u>	<u>0.0</u>
Sum	9.04	9.11	8.99	8.94	8.86	9.11	8.98
Ti+Al	7.92	7.96	8.03	8.09	8.15	7.99	8.00

Experimental conditions: Ni capsules, 5 kbar, 1 hour, temperature as shown.

Table 5. Hot-pressing of "Cs-holl. mix I" composition in graphite dies

T <sup>o</sup> C	1280 <sup>o</sup>		1250 <sup>o</sup>	
	<u>Hollandites</u>		<u>Hollandites</u>	
TiO <sub>2</sub>	66.9	69.1	66.0	67.4
Al <sub>2</sub> O <sub>3</sub>	8.8	3.8	10.5	10.1
BaO	22.1	24.1	21.4	21.7
<u>Cs<sub>2</sub>O</u>	<u>1.7</u>	<u>2.9</u>	<u>2.2</u>	<u>0.8</u>
Sum	99.5	99.9	100.1	100.0
coexisting phases	rutile corundum		rutile Al <sub>2</sub> TiO <sub>5</sub>	

Structural formulae based on 16 oxygens

Ti	6.429	6.888	6.269	6.341
Al	1.325	0.594	1.563	1.489
Ba	1.107	1.252	1.059	1.064
<u>Cs</u>	<u>0.093</u>	<u>0.164</u>	<u>0.119</u>	<u>0.043</u>
Sum	8.95	8.90	9.009	8.936
Ti + Al	7.74	7.47	7.83	7.83

Experimental conditions: graphite die, 4000 psi, 1 hour , temperature as shown.

If redox conditions are sufficiently reducing to permit the partial reduction of  $Ti^{4+}$  to  $Ti^{3+}$ , then a further solid solution mechanism is permitted, namely  $Ti^{3+} \rightleftharpoons Al^{3+}$ . It should be evident that an acceptable hollandite structural formula need not possess an integral cation sum on the basis of 16 oxygens, because of the flexibility for filling the large cation sites. However the sum of Ti + Al (all sites) should be very close to the ideal value of 8.

Hollandites produced by hot-pressing in Ni (Table 4) in most cases exhibit a modest excess of  $TiO_2$  i.e. structural  $Ti > 6$  and cation sum  $< 9.0$ . In all cases, the sum (Ti + Al) is close to a value <sup>of</sup> 8.0, and the two Al-sites are fully occupied. On the other hand, hollandites produced by hot-pressing in graphite (Table 5) have insufficient Al to <sup>substantial</sup> fully occupy the Al-site, a <sup>excess</sup> of Ti is present, and the sum of (Ti + Al) is less than the ideal value of 8. All of these characteristics are consistent with reducing conditions permitting stabilization of significant amounts of  $Ti^{3+}$ , with its subsequent entry into the Al sites. (The low sum of Ti + Al in an artifact arising from the analytical data, because the microprobe cannot discriminate between  $Ti^{3+}$  and  $Ti^{4+}$  and automatically computes <sup>all</sup> titanium as  $TiO_2$ .) The "surplus" aluminium excluded from the Al site by  $Ti^{3+}$  appears in the product assemblage as a new aluminous phase, e.g.  $Al_2TiO_5$ .

A second series of experiments was conducted to see whether  $Ti^{3+}$  would still partition into the Al-site when an excess of  $Al_2O_3$  was present. With this aim in mind a new starting composition was prepared, "Cs-holl. mix II" equivalent to  $BaAl_2Ti_6O_{16}$  (60%) +  $TiO_2$  (30%) +  $Al_2O_3$  (10%) plus an additional 2%  $Cs_2O$ . This was hot-pressed, first in a Ni capsule for comparative purposes, then in graphite, at temperatures ranging from  $1240^\circ C$  to  $1290^\circ C$  for one hour. Microprobe analyses and hollandite structural formulae are presented in Tables 6 and 7. In all cases, additional aluminous phases coexist with the product hollandites and rutile.

Table 6. Hot-pressing products of "Cs-holl. mix II"

T <sup>°</sup> C	1275 <sup>°</sup>	1290 <sup>°</sup>	1300 <sup>°</sup>	1240 <sup>°</sup>
HP container	Ni	graphite	graphite	graphite

Hollandites

TiO <sub>2</sub>	61.0	66.9	65.5	71.9	66.3
Al <sub>2</sub> O <sub>3</sub>	15.6	9.4	16.7	5.4	10.6
BaO	22.5	21.5	21.9	19.7	20.9
Cs <sub>2</sub> O	0.6	2.1	1.9	2.5	1.8
Sum	99.7	99.9	100.	99.5	99.6

Coexisting phases	rutile corundum	rutile corundum Al <sub>2</sub> TiO <sub>5</sub>	Ba-Ti-Al phase (analysis below)	Ba-Ti-Al phase (analysis below)
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TiO <sub>2</sub>	38.9	45.3
Al <sub>2</sub> O <sub>3</sub>	54.3	46.8
BaO	6.6	7.2
Cs <sub>2</sub> O	<.2	<.2
Sum	99.8	99.3

For the sample hot-pressed in Ni, corundum is an accessory. However for samples hot-pressed in graphite, a new, uncharacterised Ba-Ti-Al phase also appears in the product assemblage (Table 6). Note that cesium does not partition into this new phase. The chemical analyses and structural formulae peculiarities of the hollandites produced by hot-pressing in graphite are again consistent with stabilization of  $Ti^{3+}$  and its entry into the Al-sites. (The Ba-Ti-Al phase may also contain  $Ti^{3+}$ ). It is evident that  $Ti^{3+}$  rather than  $Al^{3+}$  preferentially enters the Al-site of hollandite, and significant partitioning of  $Ti^{3+}$  into hollandite is observed even at high  $Al_2O_3$  activities e.g. with corundum present.

Preliminary leaching data for "Cs-holl. mix II" (Table 3) are illustrated in Fig.2. These leach-rates are similar to those obtained for "Cs-holl. mix I" composition also hot-pressed in graphite. The presence of an excess of  $Al_2O_3$  has no significant effect on leaching performance.

Leaching data for formulations made up from the S03 hollandite preparation (Table 1) are also presented in Fig.2. Mixes corresponding to "S03 + 25%  $TiO_2$ " and "S03 + 20%  $TiO_2$  + 5%  $Al_2O_3$ " were hot-pressed in graphite at  $1200^\circ C$ , and "S03 + 25%  $TiO_2$ " was likewise treated at  $1300^\circ C$  and subjected to leaching at  $95^\circ C$  (Table 8). It is evident from Fig.2 that the presence or absence of  $Al_2O_3$  has no significant effect on leaching performance. It is also evident that the S03 preparation is intrinsically superior in its capacity to immobilize cesium, initial leach-rates bring almost one order of magnitude below those of "Cs-holl. mix I and II".

(b) Redox conditions controlled by addition of 2% metallic titanium

In many SYNROC formulations the addition of 2% Ti metal to the sample immediately prior to hot-pressing has been advocated as a means of ensuring that reducing conditions prevail during sample preparation.

Table 7. Structural formulae of hollandite produced in "Cs-holl. mix II"  
composition

T°C	1275 <sup>0</sup>	1290 <sup>0</sup>	1300 <sup>0</sup>	1240 <sup>0</sup>
HP container	Ni	graphite	graphite	graphite

Structural formulae based on 16 oxygen atoms

Ti	5.722	6.383	6.234	6.869	6.280
Al	2.293	1.406	1.596	0.809	1.576
Ba	1.100	1.069	1.086	0.981	1.032
<u>Cs</u>	<u>0.032</u>	<u>0.116</u>	<u>0.103</u>	<u>0.135</u>	<u>0.097</u>
Sum	9.15	8.97	9.019	8.79	8.982
Ti + Al	8.01	7.78	7.83	7.67	7.85



Table 8. Initial Cs leach-rates for S03 hollandite preparations

<u>Preparation</u>	<u>T°C</u>	<u>Day 1</u>	<u>Day 3</u>
S03 + 25% TiO <sub>2</sub>	1200	$7.2 \times 10^{-5}$	$6.7 \times 10^{-6}$
S03 + 20% TiO <sub>2</sub>	1300	$8.4 \times 10^{-5}$	$6.6 \times 10^{-6}$
S03 + 20% TiO <sub>2</sub> + 5% Al <sub>2</sub> O <sub>3</sub>	1200	$8.2 \times 10^{-5}$	$7.2 \times 10^{-6}$

Experimental conditions: hot-pressed in Ni capsules, 1 hour at temperature shown

Table 9. Effect of Ti metal on hollandite leach-rate (gm/cm<sup>2</sup> day)

<u>Hollandite preparation</u>	<u>Temp. °C</u>	<u>Without Ti metal</u>		<u>With 2% Ti metal</u>	
		<u>Day 1</u>	<u>Day 2</u>	<u>Day 1</u>	<u>Day 2</u>
S24	1200	$1.5 \times 10^{-3}$	$4.7 \times 10^{-4}$	$1.7 \times 10^{-4}$	$1.1 \times 10^{-4}$
S24	1150	$2.9 \times 10^{-3}$	$3.3 \times 10^{-4}$	$8.6 \times 10^{-5}$	$4.7 \times 10^{-5}$
S25	1200	$2.2 \times 10^{-3}$	$1.5 \times 10^{-4}$	$1.1 \times 10^{-4}$	$5.5 \times 10^{-5}$
H1	1150	$1.6 \times 10^{-3}$	$7.8 \times 10^{-5}$	$3.4 \times 10^{-5}$	$9.9 \times 10^{-6}$
H2	"	$2.3 \times 10^{-3}$	$8.3 \times 10^{-5}$	$3.2 \times 10^{-5}$	$4.8 \times 10^{-6}$
H3	"	$2.0 \times 10^{-3}$	$5.1 \times 10^{-5}$	$3.5 \times 10^{-5}$	$7.4 \times 10^{-6}$
H4	"	$1.6 \times 10^{-3}$	$3.9 \times 10^{-5}$	$4.6 \times 10^{-5}$	$5.8 \times 10^{-6}$
S24-2A	"	$7.0 \times 10^{-5}$	$6.0 \times 10^{-6}$	$5.6 \times 10^{-5}$	$2.1 \times 10^{-6}$
S25-2A	"	$1.6 \times 10^{-4}$	$2.8 \times 10^{-5}$	$2.5 \times 10^{-5}$	$1.8 \times 10^{-6}$
S16	"	$2.3 \times 10^{-3}$	$1.3 \times 10^{-4}$	$6.9 \times 10^{-4}$	$1.6 \times 10^{-4}$
S16 + 20% TiO <sub>2</sub>	"			$1.1 \times 10^{-3}$	$7.7 \times 10^{-5}$
"Cs-holl. mix 1"	1200	$1.4 \times 10^{-3}$	$2.7 \times 10^{-4}$	$1.2 \times 10^{-4}$	$3.3 \times 10^{-5}$

Experimental conditions: Samples hot-pressed 1 hour in Ni capsules.

Table 10. Hot-pressing of "Ca-holl. mix I" + 2% Ti metal in Ni capsules

T°C	1350°		1325°		1300°		1250°		1200°		1150°		1100°	
	<u>Hollandites</u>		<u>Hollandites</u>		<u>Hollandites</u>		<u>Hollandites</u>		<u>Hollandites</u>		<u>Hollandites</u>		<u>Hollandites</u>	
TiO <sub>2</sub>	66.1	67.6	71.4	74.4	66.7	68.4	71.6	68.1	74.4	74.0	71.1	72.1	70.5	72.2
Al <sub>2</sub> O <sub>3</sub>	11.1	11.1	7.1	5.7	8.4	9.0	5.8	8.1	5.8	5.9	7.0	6.1	7.1	7.3
BaO	20.7	20.6	19.8	19.7	22.1	21.7	20.3	21.5	18.7	19.4	21.0	19.9	20.0	18.9
Ca <sub>2</sub> O	2.2	0.6	1.3	0.8	2.2	1.1	2.5	2.1	2.0	0.9	2.1	1.2	2.1	1.3
Sum	101.1	99.9	99.6	100.6	99.4	100.2	100.2	99.8	100.9	100.2	101.2	99.3	99.7	99.7
coexisting phases	rutile		rutile, Al <sub>2</sub> TiO <sub>5</sub> corundum		rutile Al <sub>2</sub> TiO <sub>5</sub>		rutile corundum "Ti <sub>3</sub> O <sub>5</sub> "- type phase		rutile corundum		rutile corundum "Ti <sub>3</sub> O <sub>5</sub> "- type phase relict Ti metal		rutile corundum Al <sub>2</sub> O <sub>3</sub> -rich ss.	

Experimental conditions: Temperatures as shown above, 5 kbar, 1 hour, Ni capsules

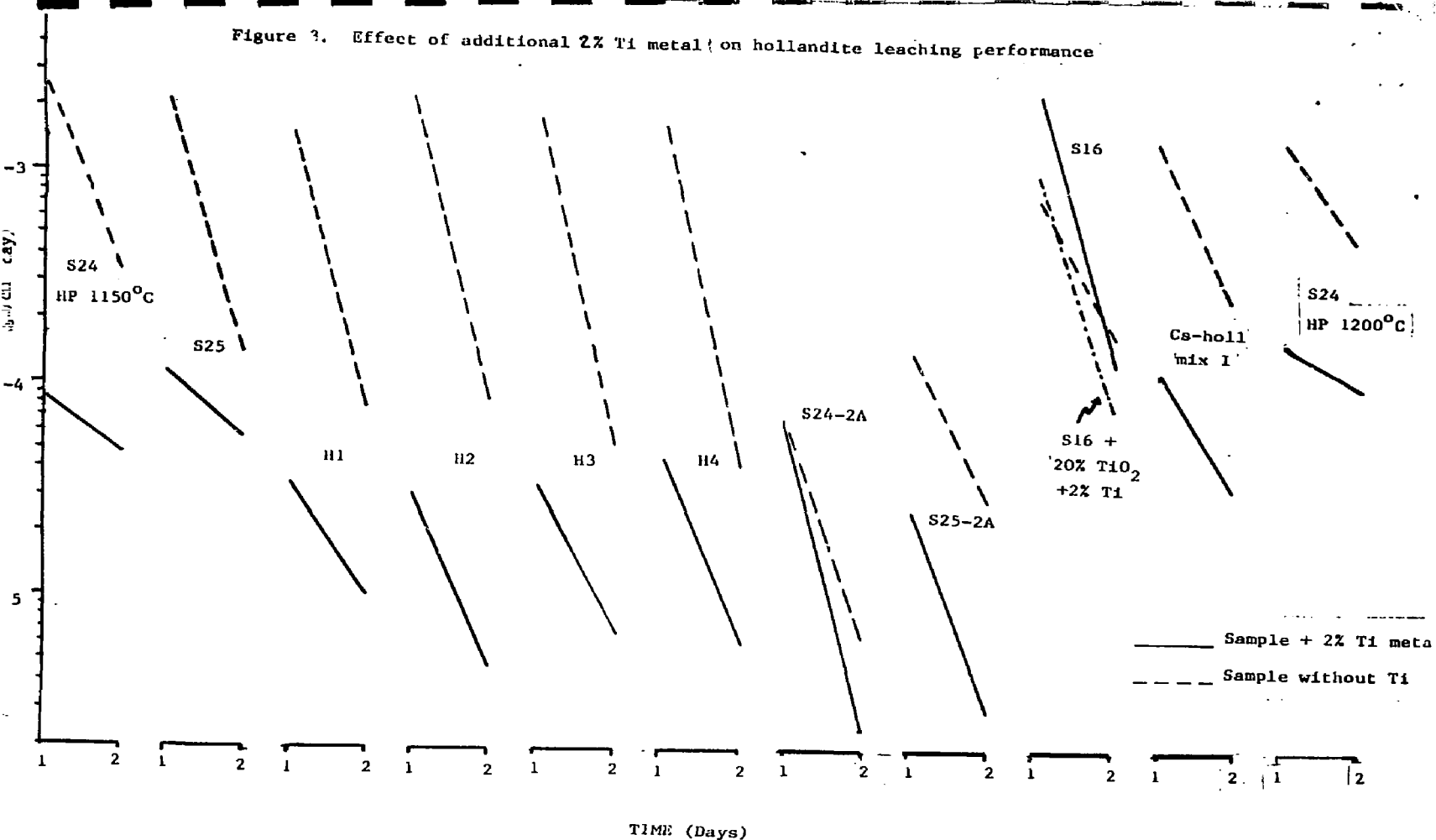
Table 11. Structural formulae of hollandite produced by hot-pressing "Cs-holl. mix I" + 2% metallic titanium

Based on 16 oxygens

T°C	<u>1350°</u>		<u>1325°</u>	<u>1300°</u>	
Ti	6.232	6.281	6.713	6.457	6.454
Al	1.640	1.616	1.046	1.274	1.331
Ba	1.017	0.997	0.970	1.115	1.069
Cs	0.118	0.032	0.069	0.121	0.059
Sum	9.007	8.927	8.799	8.967	8.910
Ti + Al	7.87	7.89	7.76	7.68	7.78

T°C	<u>1250°</u>		<u>1200°</u>		<u>1150°</u>		<u>1100°</u>	
Ti	6.814	6.523	6.891	6.874	6.705	6.816	6.686	6.725
Al	0.865	1.216	0.842	0.859	1.035	0.904	1.055	1.066
Ba	1.007	1.073	0.903	0.939	0.983	0.980	0.988	0.917
Cs	0.135	0.114	0.105	0.047	0.112	0.064	0.113	0.068
Sum	8.821	8.926	8.741	8.720	8.834	8.764	8.843	8.777
Ti + Al	7.68	7.73	7.71	7.73	7.74	7.72	7.74	7.79

Figure 3. Effect of additional 2% Ti metal on hollandite leaching performance



The metallic titanium serves to scavenge oxygen either from trapped air or residual nitrates, and may even scavenge nitrogen. In many cases relict Ti metal can be found mantled with  $TiO_2$ - $Ti_3O_5$  reaction products, indicating that (at least locally) redox conditions are sufficiently reducing so that some  $Ti^{3+}$  as well as  $Ti^{4+}$  are stable species.

The leaching performance of hollandite improves markedly if ~ 2% Ti metal is incorporated prior to hot-pressing. The ten preparations used in this study are summarized in Table 1. Table 9 and Fig.3 illustrate how Cs-based leach-rates fall by about one order of magnitude after the addition of Ti-metal, irrespective of the nature of the starting material.

Microprobe investigations again were focussed on the "Cs-holl. mix I" composition. It is evident from the microprobe analyses presented in Table 10 and the structural formulae of Table 11 that the effect of Ti metal is to lower experimental redox conditions so that  $Ti^{3+}$  is stabilized, and enters the Al-site of hollandite. As in the case where this same starting material was hot-pressed in graphite (Table 5) the hollandites exhibit an "excess" of Ti ( $> 6$ ), a deficiency in Al ( $< 2$ ) and an artificially low value for the sum of Ti + Al. Moreover the product assemblages contain an additional aluminous phase such as corundum or pseudobrookite ( $Al_2TiO_5$  -  $Ti_3O_5$  solid solutions). This is consistent with displacement of  $Al^{3+}$  from hollandite by  $Ti^{3+}$ . There is no correlation between the  $Cs_2O$  content of hollandite and its "Al-deficiency" i.e. the amount of  $Ti^{3+}$  substitution. In other words, there is unfortunately no evidence for the fixation of Cs in hollandite via the specific exchange mechanism of  $2Cs^+ + Ti^{3+}$  for  $Ba^{2+} + Al^{3+}$ .

#### (c) Leaching behaviour of S16

The leaching behaviour of hollandite preparation S16 stands out sharp contrast to the remainder of the samples in Fig.3. There is no significant difference in leaching performance when 2% Ti metal is added

to S16 alone, or to S16 + 20%  $\text{TiO}_2$ . This anomalous behaviour is believed to arise because S16 is a "stoichiometric" hollandite with molar  $(\text{BaO} + \text{Cs}_2\text{O}) : (\text{Al}_2\text{O}_3) : (\text{TiO}_2) = 1:1:6$  with no solid solution towards the  $\text{Ba}_{10}\text{Al}_{10}\text{Ti}_{18}\text{O}_{86}$  endmember. Reaction of Ti metal with " $\text{TiO}_2$ " from the hollandite produces  $\text{Ti}^{3+}$ , which in turn displaces some Al from the Al-site. This reaction in effect shifts the bulk composition to molar proportions of  $1 : (1+x) : (6-y)$ . Clearly this cannot crystallize as a single-phase hollandite. Any new phases which are thus stabilized may preferentially incorporate cesium and may also be relatively easily leached. This behaviour may account for the poor leaching performance of S16 + 2% Ti (Fig.3).

In principle, the addition of 20%  $\text{TiO}_2$  to this latter composition should significantly improve its leaching performance, because there is more than enough  $\text{TiO}_2$  to react with Ti metal, and the modification of the bulk composition as outlined above, will not occur. However as Fig.3 indicates, the leaching behaviour of S16 + 20%  $\text{TiO}_2$  + 2% Ti is not significantly different. This disappointing result is attributed to problems arising from hand-mixing  $\text{TiO}_2$  (anatase) with the spray-dried S16 preparation. Flocculation of anatase may have prevented satisfactory equilibration during hot-pressing, thus the problems encountered with the previous preparation were not eliminated.

##### 5. Long-term leaching of hollandite

Ringwood et al. (1981b) reported leaching data for two hollandite preparations for periods to 60 days; by which time Cs-based leach-rates had fallen to between  $5 \times 10^{-7}$  and  $1 \times 10^{-6} \text{ gm/cm}^2 \text{ day}$ . Both samples had been hot-pressed in graphite and were low-Al,  $\text{Ti}^{3+}$ -bearing hollandites similar to those described in this report.

Further long-term leaching has been conducted on the Sandia Labs. preparation S25-2A, a hollandite already remarkable for a low initial leach-rate. Its leaching performance (Table 12) is illustrated in

Table 12: Long-term leaching of hollandite S25-2A

<u>Day</u>	<u>leach-rate g/cm<sup>2</sup> . day</u>
1	$3.8 \times 10^{-5}$
2-3	$7.1 \times 10^{-6}$
4-6	$2.5 \times 10^{-6}$
7-10	$1.2 \times 10^{-6}$
11-17	$8.3 \times 10^{-7}$
18-27	$4.7 \times 10^{-7}$
28-49	$2.4 \times 10^{-7}$
50-69	$2.1 \times 10^{-7}$
70-99	$1.2 \times 10^{-7}$

Experimental conditions: Graphite die, 1265°C, 1 hour, 4000 psi.



Figure 4. Long-term leaching of S25-2A hollandite

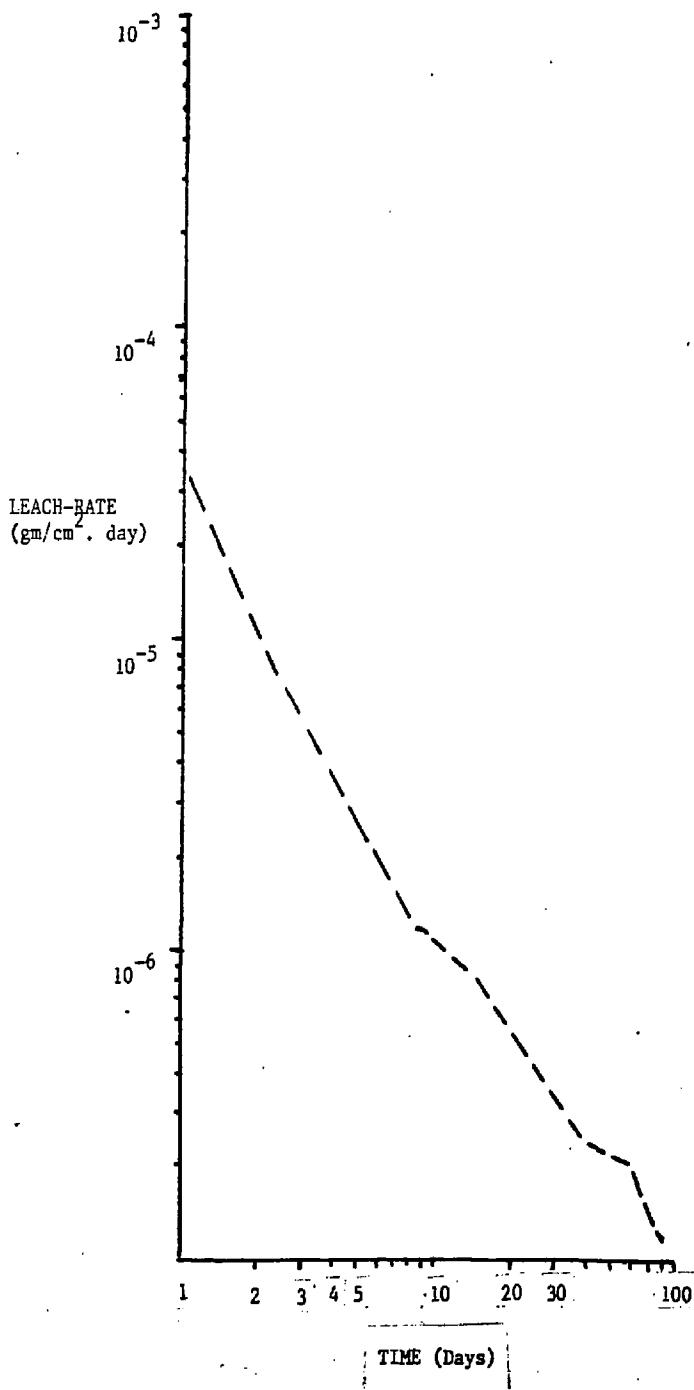


Fig.4. Note that after 10 days, its leach-rate has fallen to a new low of  $1.2 \times 10^{-7}$  gm/cm<sup>2</sup>. day. So far, there is no evidence for the "plateau" reported in some of our earlier SYNROC leaching studies.

## 6. Conclusions

Hollandites prepared by the Sandia method exhibit superior leaching behaviour in comparison to hollandites prepared by any other technique. The leaching performance of all hollandites irrespective of preparation technique, can be markedly improved by hot-pressing under specific controlled redox conditions, below Ni-NiO. These conditions can be achieved by hot-pressing in graphite dies at modest confining pressures, or by the addition of 2% finely dispersed Ti metal powder to the sample prior to hot-pressing. For large-volume samples, the latter method will yield the best results. Hollandites produced under these conditions are characterised by an "excess" of TiO<sub>2</sub> and a "deficiency" in Al<sub>2</sub>O<sub>3</sub>. In effect, Ti<sup>3+</sup> stabilized at these ambient low oxygen fugacities, preferentially partitions into the Al-site, displacing Al<sup>3+</sup> into new aluminous phases such as corundum or pseudobrookite.

There is no satisfactory crystal-chemical reason why the leaching performance of Ti<sup>3+</sup>-bearing hollandite should be superior to that of Al<sup>3+</sup> hollandite. The most likely explanation is that there are kinetic factors involved and reaction pathways may be different when Ti<sup>3+</sup> is present, since the activity of (Ti + Al)<sup>3+</sup> is necessarily higher in these cases. Metastable Cs-bearing phases formed prior to or during hollandite nucleation may not be completely eliminated unless Ti<sup>3+</sup> is present in the system. These relict metastable phases may contribute to poor Cs leach-rates in hollandite formulations hot-pressed in Ni, and without additional Ti metal.

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# Incorporation of Uranium and Rare Earths into Zirconolite

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## 1. Introduction

The rare earths and trivalent and tetravalent actinides are immobilized in SYNROC by two phases, zirconolite and perovskite. A reconnaissance experimental study has been undertaken to explore the tolerance of the zirconolite lattice to substitutions involving the above elements, since high concentrations of actinides or rare earths may tend to stabilize a less-desirable pyrochlore phase in preference to zirconolite. The experiments were specifically aimed at determining how much uranium, and samarium or neodymium could be accommodated in the zirconolite lattice before it was destabilized in favour of pyrochlore. Here, uranium serves as a model for all tetravalent actinides and the rare earths also model the behaviour of the trivalent actinides. Naturally-occurring zirconolites contain as much as 23 wt.%  $\text{ThO}_2 + \text{UO}_2$  and up to 9% rare earths (REE). Sinclair and Ringwood (1981) have shown that these samples, which are now metamict, recrystallize to recover their zirconolite structure after heat-treatment at  $\sim 1100^\circ\text{C}$ .

## 2. Paragenesis of Zirconolite

Zirconolite occurs in trace amounts in a variety of environments and is usually found as small grains up to several millimeters in dimension. At Jacupiranga, Brazil, it occurs in the magnetite-pyroxenite rock which intrudes the carbonatite complex (Hussak and Prior, 1895). In Germany, zirconolite has been found in several dikes which intrude the Kaiserstuhl carbonatite (Keller, pers. comm.).

The mineral has been reported in several locations in the USSR (Vlasov, 1966). Samples from Arbarastakh in Aldan are contained in the central regions of pyroxenite massifs. The

pyroxenites outcrop in a group of ultrabasic alkali rocks and have been subjected to intense metasomatic alteration (Borodin et al., 1960). A niobium-rich variety of zirconolite from the apatite-magnetite rocks of the Vuoriyarvi massif has also been reported by these authors. Zirconolite from the Kola Peninsula occurs in the coarse grained pyroxene-amphibole rock of the Afrikanda massif.

In Sri Lanka, zirconolite occurs as large pebbles weighing up to 60 grams. They are found in gem gravels at several localities (Blake and Smith, 1913).

The oldest known specimens on earth have been found in the Jimberlana Intrusion, Western Australia. The intrusion consists of a layered horizontal mafic complex and has been dated at about 2500 m.y. old (Turek, 1966; Oversby and Ringwood, 1980). Recently, Williams (1978) has reported an occurrence of zirconolite in the Rhum ultrabasic layered intrusion. The zirconolite is interpreted to be a late stage mineral containing the remaining incompatible elements of the original magma.

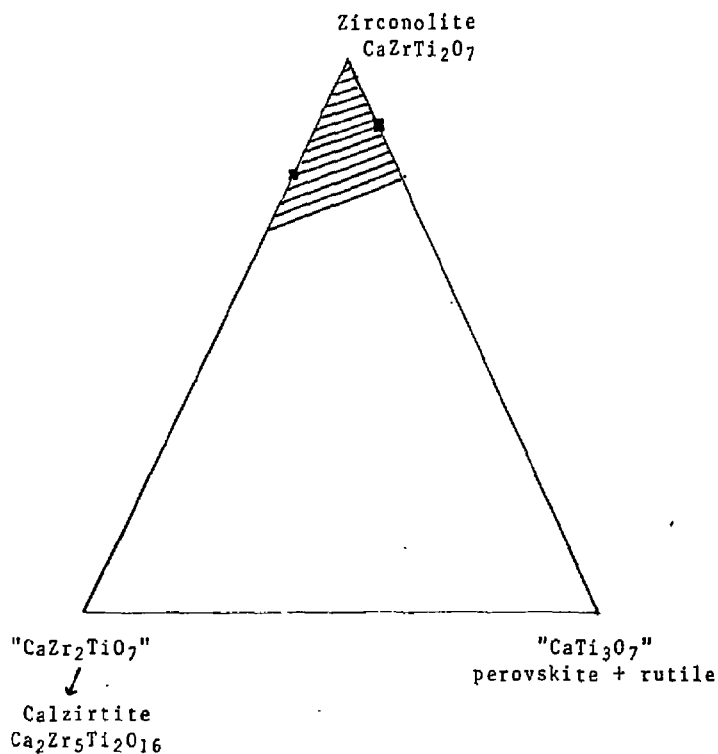
Samples from Norway are known as polymignite (Palache et al., 1944; Vlasov, 1966). This mineral is found in Norway in a pegmatite with Na-orthoclase, magnetite, nepheline, zircon and pyrochlore. It has also been found in augite-syenite rocks.

### 3. Crystal Chemistry of Zirconolite

The crystal structure of zirconolite has been described in detail by Rossell (1980a) and Gatehouse et al. (1981). They confirmed earlier work by Pyatenko and Pudovinka (1961) by showing that it is related to structures of the defect fluorite-type, in that it is a superstructure derived by ordering of  $\text{CaF}_{2-x}$  type sub-units. The centred monoclinic unit cell of zirconolite is eight times as large as the fluorite-type sub-unit, and its structure is more correctly expressed by the formula  $\text{Ca}_8\text{Zr}_8\text{Ti}_{16}\text{O}_{56}\square_8$  ( $\square$  symbolises an oxygen vacancy).

Rossell (1980a) found that the Ca atoms are coordinated to 8 oxygens at the corners of a distorted cube. Zirconium, in seven-fold coordination, is linked to 7 oxygens at the corners of a truncated cube. Titanium occupies three distinct lattice sites. In two of these, it is in 6-fold coordination, linked to

Fig. 1. The system  $\text{CaZrTi}_2\text{O}_7$  - " $\text{CaTi}_3\text{O}_7$ " -  $\text{CaZr}_2\text{TiO}_7$ .



Shaded field shows limits of zirconolite solid solution towards calzirtite and " $\text{CaTi}_3\text{O}_7$ " as defined by Gatehouse et al. (1980). Solid squares indicate compositions synthesised at  $1350^\circ\text{C}$  and  $1400^\circ\text{C}$  at ANU.

6 oxygens at the vertices of octahedra. These octahedra are linked at their corners, forming either 6-membered or 3-membered rings. The third type of titanium site involves 5-fold coordination with titanium linked to 5 oxygens at the apices of a trigonal bipyramid.

Crystal-chemical studies of zirconolite by Gatehouse et al. (1981) have defined the compositional range of zirconolite. They also showed that the lattice parameters range systematically as a function of composition and temperature. They found that zirconolite was remarkably intolerant to variations in its Ca content, permitting changes of only a few atom %. However, considerable variation in Zr/Ti ratios is permitted. If the formula of zirconolite is written as  $\text{CaZr}_x\text{Ti}_{3-x}\text{O}_7$  then  $x$  ranges from 0.9 to 1.3. In other words, zirconolite exhibits solid solutions towards two other components, namely  $\text{Ca}_2\text{Zr}_5\text{Ti}_2\text{O}_{16}$  (calzirtite) and " $\text{CaTi}_3\text{O}_7$ " (Fig. 1). Experiments in our laboratory have confirmed that at  $1350^\circ\text{C}$ , zirconolite solid solutions with up to 10 mole % of " $\text{CaTi}_3\text{O}_7$ " are stable. This is relevant to SYNROC formulations where zirconolite coexists with  $\text{CaTiO}_3 \pm$  rutile at similar temperatures. Similar experiments have established that solid solutions with 18 mole %  $\text{CaZr}_2\text{TiO}_7$  component can be synthesized at  $1400^\circ\text{C}$ . These results are shown in Figure 1.

Natural and synthetic zirconolites have been found to accommodate a wide range of cation substitutions. In order to facilitate discussion of the crystal chemistry we will simply describe cation substitutions in terms of the Ca- Zr- or Ti-sites. Table 1 presents microprobe analyses of a suite of natural zirconolites from many countries. Of particular relevance to radwaste immobilization is the observation that natural zirconolites may contain as much as 23 wt.%  $\text{ThO}_2$  plus  $\text{UO}_2$ , and 9 wt.% rare earth elements (Table 1). Because many natural zirconolites are metamict, the distribution of cations amongst the various sites cannot be resolved by conventional X-ray techniques. However, the structural formulae of the natural samples (Table 1) when coupled with crystal chemical considerations, are quite revealing. The 8-fold coordinated Ca-site should be particularly suited for large cations e.g.  $\text{Ca}^{2+}$ ,  $\text{U}^{4+}$ ,  $\text{Th}^{4+}$  and the larger rare earths. The 7-fold coordinated Zr-site would best be occupied by  $\text{Zr}^{4+}$ , some

Table 1. Chemical analyses of natural ZIRCONOLITES.

Locality	Stavern (Brown)	Stavern (Red)	Larvik	Jacupiranga	Aldan	Kaiserstuhl	Sri Lanka (B20392)	Sri Lanka (83800)	Sri Lanka (SL3-12)
Nb <sub>2</sub> O <sub>5</sub>	18.9	15.6	8.0	11.3	4.6	15.6	3.6	3.9	<0.2
Ta <sub>2</sub> O <sub>5</sub>	0.4	<0.2	0.7	0.2	<0.1	1.2	-	-	<0.1
TiO <sub>2</sub>	16.7	22.4	27.0	24.3	35.4	20.9	28.0	27.0	36.5
ZrO <sub>2</sub>	32.0	35.3	31.4	35.2	36.5	34.8	30.9	30.4	38.6
ThO <sub>2</sub>	3.7	0.6	4.1	7.4	1.2	2.9	20.5	21.8	0.9
UO <sub>2</sub>	1.9	0.3	1.3	0.5	0.5	1.5	2.1	1.7	4.6
FeO*	9.3	8.4	6.9	7.2	5.0	7.8	3.1	3.0	3.4
MnO	.5	0.4	0.5	0.2	<0.1	0.2	-	0.2	0.2
MgO	-	-	-	0.4	-	0.3	2.4	2.5	0.6
CaO	7.7	10.0	9.2	11.5	15.1	12.1	7.9	7.7	12.1
REE,Y	6.6	4.8	9.3	2.6	1.9	0.9	-	-	-
Sum	97.7	97.8	98.4	100.8	99.2	98.2	98.5	97.2	96.9

Electron microprobe analyses by N.G. Ware, except data for Larvik (A. Raheim, pers. comm.). \*All iron assumed to be FeO.

#### Structural formulae based on 7 oxygen atoms

Nb	0.596	0.463	0.243	0.332	0.126	0.464	0.113	0.125	-
Ta	0.008	-	0.013	0.004	-	0.022	-	-	-
Ti	0.876	1.105	1.366	1.186	1.611	1.034	1.456	1.433	1.718
Zr	1.089	1.129	1.030	1.114	1.077	1.117	1.042	1.047	1.178
Th	0.059	0.009	0.063	0.109	0.003	0.043	0.323	0.350	0.013
U	0.030	0.004	0.020	0.007	0.006	0.022	0.032	0.027	0.064
Fe"	0.543	0.461	0.388	0.391	0.253	0.429	0.179	0.118	0.178
Mn	0.030	0.022	0.029	0.011	-	0.011	-	0.012	0.011
Mg	-	-	-	0.039	-	0.029	0.247	0.263	0.056
Ca	0.576	0.703	0.663	0.800	0.978	0.853	0.585	0.582	0.811
REE,Y	0.159	0.109	0.216	0.058	0.040	0.020	-	-	-
Sum	3.963	4.004	4.030	4.051	4.094	4.045	3.978	3.957	4.028



Ti<sup>4+</sup> and the smaller rare earths. In section 4 of this paper we present evidence for the somewhat unexpected substitution of U<sup>4+</sup> into this site. The Ti-sites, in 6- and 5-fold coordination, should be particularly suited for small cations such as Ti<sup>4+</sup>, Nb<sup>5+</sup>, Ta<sup>5+</sup>, Mg<sup>2+</sup>, Mn<sup>2+</sup>, Fe<sup>2+</sup>, Fe<sup>3+</sup> and minor Zr<sup>4+</sup>.

Inspection of the structural formulae of the natural samples (Table 1) reveals some interesting trends. With one exception (Aldan), full occupancy of the Ca-site requires significant amounts of other cations as well as calcium. Even if all the available Th and U is then allocated to the Ca-site, it is still not fully occupied. Small amounts of rare earths are also required. Note that even if all rare earths were fully partitioned into the Ca site, some U and Th are nevertheless required for full occupancy. As discussed above, the Zr site can contain small amounts of Ti, whilst small amounts of Zr may partition into the Ti-sites. Whilst we cannot deduce the exact nature of this partitioning from the structural formulae, it is nevertheless significant that all samples contain in excess of 1.00 Zr atoms and in many cases, the amounts of Nb, Ta, Mg, Mn, Fe and Ti are insufficient to fill the two Ti-sites completely. It follows that a significant proportion of Zr must be partitioned into the Ti sites. The above constraints on site occupancy also lead to the conclusion that most Th and U is concentrated in the Ca-site, with very little being partitioned into the Zr-site. In a study of synthetic zirconolites, Rossell (1980b) came to a similar conclusion.

Zirconolites which crystallize in SYNROC formulations aimed at immobilizing US defense wastes contain significant amounts of other elements such as Mn, Fe, Al and minor amounts of Na and Ni. Iron is probably present as both divalent and trivalent cations, so there are uncertainties introduced in the derivation of structural formulae from microprobe analyses where iron is automatically calculated as "FeO". However, some trends are significant - a covariance between CaO and MnO suggests Mn<sup>2+</sup> is strongly partitioned into the Ca-site. Strontium is most reluctant to enter zirconolite under any circumstances, indicating that the Ca-site cannot readily accommodate ions much larger than Ca<sup>2+</sup>, whereas smaller species (e.g. Mn<sup>2+</sup>) fit comfortably. Nickel contents of zirconolites are invariably low (typically less than 0.5%) suggesting that the octahedrally-coordinated Ti-sites are suitable only

for smaller cation species. The site occupancy of iron is uncertain, but it seems reasonable to believe it could occupy both Zr- and Ti-sites. The preferred site occupancy of certain key elements such as Mg, Fe, Co, Ti and rare earths should soon be confirmed by the results of a detailed study presently being carried out by Rossell (pers. comm.).

#### 4. Incorporation of Uranium into Zirconolite

The solid solution limits between  $\text{CaZrTi}_2\text{O}_7$  (zirconolite) and  $\text{CaUTi}_2\text{O}_7$  (pyrochlore) were explored in order to establish how much substitution of zirconium by uranium could be tolerated by the zirconolite structure before it was destabilized in favour of pyrochlore. Compositions corresponding to  $\text{CaZr}_{.25}\text{U}_{.75}\text{Ti}_2\text{O}_7$ ,  $\text{CaZr}_{.5}\text{U}_{.5}\text{Ti}_2\text{O}_7$  and  $\text{CaZr}_{.75}\text{U}_{.25}\text{Ti}_2\text{O}_7$  were hot-pressed at  $1400^\circ\text{C}$ . Trace amounts of  $(\text{Zr,Ti,U})\text{O}_2$  solid solutions were observed in all runs. Homogeneous pyrochlore phases were synthesized from the first two starting materials. Product assemblages were studied by X-ray diffraction and electronprobe microanalysis. The  $\text{CaZr}_{.75}\text{U}_{.25}\text{Ti}_2\text{O}_7$  preparation yielded a pyrochlore + zirconolite assemblage. The compositions and structural formulae of this pair are given in Table 2. Rossell (1980b) has argued that the entry of Th (and by implication, U) into the zirconolite lattice involves its partition into the Ca site, with coupled substitution of a lower-valency ion (e.g.  $\text{Ti}^{3+}$ ) into the Zr site. Whilst this may well have occurred in his experiments, we find no evidence for substitution mechanisms of this kind in our own experiments. The structural formulae in Table 2 indicate that the entry of uranium into the zirconolite lattice is via a simple substitution.  $\text{U}^{4+}$  proxies primarily for  $\text{Zr}^{4+}$ , with a very small amount also being partitioned into the Ti sites. The  $\text{UO}_2$  content of this zirconolite is 13%. This type of behaviour has been observed many times in other SYNROC formulations.

It is possible that the entry of  $\text{U}^{4+}$  into the Ca-site might be facilitated by a coupled substitution whereby  $\text{U}^{4+}$  replaces  $\text{Ca}^{2+}$  and a lower-valence cation (e.g.  $\text{Mg}^{2+}$ ) substitutes for  $\text{Ti}^{4+}$ . To this end, we investigated the solid solution limits between U-bearing zirconolite and pyrochlore and the hypothetical end-member " $\text{UZrTiMgO}_7$ ". The product assemblages contained various  $(\text{Ti,Zr,U})\text{O}_2$  solid solutions and some uncharacterized phases,

Table 2. Coexisting pyrochlore and zirconolite synthesized from  $\text{CaU}_{.25}\text{Zr}_{.75}\text{Ti}_2\text{O}_7$ .

	<u>Zirconolite</u>	<u>Pyrochlore</u>
$\text{TiO}_2$	42.3	37.3
$\text{ZrO}_2$	29.1	18.9
$\text{UO}_2$	12.6	28.0
$\text{CaO}$	15.2	15.7
Sum	<u>99.2</u>	<u>99.9</u>

Structural formulae based on 7 oxygen atoms.

Ti	1.955	1.891
Zr	0.872	0.621
U	0.172	0.420
Ca	1.001	1.134
Sum	<u>4.001</u>	<u>4.067</u>

Zirconolite

Model molecule:  $\text{Ca}_{1.00}(\text{U},\text{Zr})_{1.00}(\text{Ti},\text{Zr})_{2.00}\text{O}_7$

Starting composition:  $\text{CaU}_{.25}\text{Zr}_{.75}\text{Ti}_2\text{O}_7$

Experimental conditions: 5 kbar,  $1400^\circ\text{C}$ , 1 hr.,  
Pt capsule.

however, one phase, with a very narrow composition range, was common to all product assemblages. Its composition range and structural formula are presented in Table 3. X-ray diffraction established that this phase is a zirconolite. Note how the entry of Mg into the lattice has permitted substitution of additional  $U^{4+}$ . The  $UO_2$  content of this zirconolite (~26%) is substantially higher than that (13%) of the Mg-free zirconolite in Table 2. The structural formulae and "model molecules" calculated on the basis of our analytical data confirm that additional  $U^{4+}$  can be persuaded to enter the Ca-site of the zirconolite lattice by means of a coupled substitution involving  $Mg^{2+}$  entering the Ti-sites.

Rossell (1980b) has proposed that  $Ti^{3+}$  (or  $Mg^{2+}$ ) enter the Zr-site, thereby allowing  $Th^{4+}$  (and hence  $U^{4+}$ ) to enter the Ca-site. We find it difficult to accept that  $Ti^{3+}$  was a stable species in his experiments, since most of these apparently involved synthesis in air. (Other work in our laboratory finds evidence for  $Ti^{3+}$  only when oxygen fugacity conditions are very low indeed, below the Fe-FeO buffer). Moreover, we suspect that small cations would prefer the 5- and 6-fold coordinated Ti-sites. However, Rossell's data are actually reasonably consistent with our own preferred interpretation of site occupancies, namely that (Ca,U) (Zr,U,Ti) and (Ti,Mg,Zr) respectively occupy the Ca, Zr, and Ti-sites of zirconolite. All the Ti in the Zr-site is in the 4+ state; there is no evidence to suggest that  $Ti^{3+}$  is stabilized.

In Section 3 we concluded that in natural zirconolites a significant proportion of uranium was located in the Ca-site. This probably reflects a coupled substitution of the above type, involving  $Mn^{2+}$ ,  $Fe^{2+}$  and  $Fe^{3+}$  as well as  $Mg^{2+}$ .

##### 5. Incorporation of Rare Earths into the Zirconolite Lattice

A starting mixture with composition corresponding to 1:1 molar  $CaZrTi_2O_7$  and  $NdYTi_2O_7$  was hot-pressed at  $1500^{\circ}C$  for 1 hr. Neodymium and yttrium are similar in size to calcium and zirconium respectively and so it was anticipated that the zirconolite lattice would be especially tolerant to the coupled substitution of Nd + Y for Ca and Zr (yttrium behaves as an excellent crystal-chemical

Table 3. Mg-bearing U-rich zirconolites.

TiO <sub>2</sub>	33.4	34.6
ZrO <sub>2</sub>	25.1	25.1
UO <sub>2</sub>	26.8	25.1
MgO	1.6	2.4
CaO	<u>13.2</u>	<u>12.8</u>
Sum	<u>100.1</u>	<u>100.0</u>

Structural formulae based on 7 oxygen atoms

Ti	1.704	1.735
Zr	0.831	0.816
U	0.405	0.372
Mg	0.162	0.239
Ca	<u>0.960</u>	<u>0.915</u>
Sum	<u>4.061</u>	<u>4.077</u>

Model molecules<sup>1</sup> (Ca,U)<sub>1.00</sub>(Zr,U)<sub>1.00</sub>(Ti,Mg,Zr)<sub>2.06</sub>O<sub>7</sub>  
 (Ca,U)<sub>1.00</sub>(Zr,U)<sub>1.00</sub>(Ti,Mg,Zr)<sub>2.08</sub>O<sub>7</sub>

Experimental conditions: 5 kbar, 1400°C, 1 hr., Pt capsule.

Starting compositions: Ca(Zr,U)Ti<sub>2</sub>O<sub>7</sub> ss. + "UZrMgTiO<sub>7</sub>"

<sup>1</sup>Calculated by filling Ca-site with Ca+U, then filling Zr-site with Zr+U. Residual Zr plus Mg and Ti are allocated to Ti-sites.

analogue for the heavy rare earths). However, the experimental product turned out to be a homogeneous single-phase pyrochlore solid solution whose composition is given in Column 1, Table 4.

A second composition corresponding to a 1:1 molar mixture of two end-member phases,  $\text{CaZrTi}_2\text{O}_7$  and  $\text{Sm}_2\text{Ti}_2\text{O}_5$  was hot-pressed at  $1400^\circ\text{C}$  and  $1450^\circ\text{C}$ . In this instance, a pair of coexisting phases (low-Zr and high-Zr phases of Table 2) were produced. Subsequent synthesis of these two phases confirmed that the low-Zr phase is a pyrochlore, whilst the high-Zr phase is a zirconolite. (The composition of the newly-synthesized high-Zr phase is given under "C" in Table 4. Note the high rare earth content of 27%  $\text{Sm}_2\text{O}_3$ ). However, because the starting material consisted of a mixture of two previously-synthesized end-members, we recognised the possibility that the coexisting pyrochlore-zirconolite pair might be metastable with respect to a homogeneous single phase. This was confirmed when a new starting mix of oxides + carbonates corresponding to the appropriate composition was hot-pressed at  $1450^\circ\text{C}$  and a single phase pyrochlore-type (f.c.c.) solid solution was produced.

## 6. Conclusions

Electron microprobe analyses of a suite of natural zirconolites reveal that Nb, Ta, Th, U, Fe, Mn, Mg, Y and rare earths can be accommodated in the zirconolite lattice. Many of the natural samples are metamict but have recrystallized to single-phase zirconolite after heat treatment (Sinclair and Ringwood, 1981). In SYNROC formulations, and along the binary join  $\text{CaZrTi}_2\text{O}_7$ - $\text{CaUTi}_2\text{O}_7$ , entry of U into the zirconolite lattice essentially involves substitution in the Zr-site. Natural zirconolites, on the other hand, appear to accommodate significant amounts of U in the 8-fold coordinated Ca-site. Presumably, in chemically-complex natural minerals, this substitution is facilitated by entry of small, lower valence cations into the Ti-sites e.g.  $\text{Al}^{3+}$ ,  $\text{Fe}^{3+}$ . Laboratory experiments have confirmed that the entry of U into the Ca-site is facilitated by a coupled substitution whereby a small divalent cations such as  $\text{Mg}^{2+}$  is simultaneously partitioned into the Ti-sites. Natural zirconolites may contain up to 23%  $\text{ThO}_2 + \text{UO}_2$ ; synthetic zirconolites along the  $\text{CaZrTi}_2\text{O}_7$  -  $\text{CaUTi}_2\text{O}_7$  join take up only ~13%  $\text{UO}_2$ . (At higher  $\text{UO}_2$  contents a pyrochlore phase is preferentially stabilized). However, when coupled substitutions

Table 4. Rare earth-bearing pyrochlores and zirconolite.

	A	B		C	D
		Low Zr-phase	High-Zr phase		
TiO <sub>2</sub>	40.8	38.4	40.0	40.5	37.6
ZrO <sub>2</sub>	16.0	14.0	22.0	24.9	16.3
Sm <sub>2</sub> O <sub>3</sub>	-	40.5	29.4	26.5	38.5
Nd <sub>2</sub> O <sub>3</sub>	21.5	-	-	-	-
Y <sub>2</sub> O <sub>3</sub>	14.	-	-	-	-
MgO*	0.4	0.8	0.6	-	SrO=0.4
CaO	7.2	6.1	7.7	8.5	7.2
Sum	99.9	99.8	99.9	100.4	103.0

Structure: pyrochlore pyrochlore zirconolite zirconolite pyrochlore-type

Structural formula based on 7 oxygens:

Ti	1.974
Zr	0.787
Sm	0.592
Ca	0.590
Sum	<u>3.943</u>

Starting composition: A =  $\text{CaZrTi}_2\text{O}_7$  +  $\text{NdYTi}_2\text{O}_7$  1:1 molar  
 B =  $\text{CaZrTi}_2\text{O}_7$  +  $\text{Sm}_2\text{Ti}_2\text{O}_7$  1:1 molar mixture of the  
 two components  
 C = High-Zr phase from "B"  
 D =  $\text{CaZrTi}_2\text{O}_7$  +  $\text{Sm}_2\text{Ti}_2\text{O}_7$  oxide/carbonate mix.

Experimental conditions: 5 kbar, 1450°C, 1 hr., Pt capsule.

involving Mg takes place,  $UO_2$  contents of 27% can be achieved.

SYNROC formulations developed for immobilizing civilian high-level wastes typically comprise about 35 wt.% zirconolite. Waste-loadings may be as high as 20% and the calcined waste itself contains about 1%  $UO_2$ . From these constraints we deduce that the zirconolite will be required to accommodate only about 0.6%  $UO_2$ , and will be way below its "saturation limit". The situation is somewhat different in the case of SYNROC formulations designed to immobilize the iron and alumina-rich sludges resulting from the US Defense program. These calcined sludges contain quite high concentrations of  $UO_2$  (~3%). Moreover, the waste-loading of SYNROC is now much higher (~65%) and the wasteform itself contains a smaller proportion of zirconolite (~13%). It follows that the zirconolite phase will be required to immobilize about 15 wt.%  $UO_2$ . If this uranium were to be accommodated by simple substitution into the Zr-site, the zirconolite phase would be dangerously close to the "saturation limit". However, the high concentrations of iron, aluminium etc. in the Defense sludges give rise to zirconolites which themselves contain significant amounts of these elements. The zirconolite is thus able to accommodate additional uranium via the coupled substitution mechanism discussed earlier, and is still well below its "saturation limit" with respect to uranium and tetravalent actinides.

Natural zirconolites may contain at least as much as 9% rare earths. Zirconolites in SYNROC formulations will be required to accommodate only a few percent of rare earths and trivalent actinides. Experiments have shown that the zirconolite lattice can accept up to 30% rare earths ( $Sm_2O_3$ ) before becoming destabilized in favour of pyrochlore or a related f.c.c. structure. SYNROC zirconolites will therefore be well below their "saturation limits" in rare earths and trivalent actinides.

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RECONNAISSANCE STUDIES OF THE STABILITY RELATIONS  
OF Ca-Ti-Al PHASES IN SYNROC-C FORMULATIONS

S.E. Kesson and A.E. Ringwood

1. Introduction

Ca-Ti-Al bearing phases have been observed in many SYNROC-C preparations. X-ray data indicate two distinct structural types (i) a previously uncharacterised Ca-Ti-Al phase (CTA) with the composition  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$  and (ii) Fe,Ti-bearing calcium aluminate phases with the magnetoplumbite structure. A reconnaissance experimental program has been undertaken in order to characterise these phases and to elucidate their stability relations in SYNROC-C formulations.

The interpretation of experimental assemblages in this program was beset by many difficulties. For example, corundum and perovskite, once nucleated, are notoriously reluctant to react again outside their stability fields. Secondly, in many simple systems (e.g.  $\text{Al}_2\text{O}_3$ - $\text{TiO}_2$ - $\text{CaO}$ ), melting occurred, as evidenced by abrupt textural changes (e.g. development of euhedral phenocrysts). The liquid always quenched to a multiphase crystalline matrix, and so it follows that phenocryst + matrix phases do not necessarily represent an equilibrium assemblage. Finally, the interpretation of subsolidus experiments was hampered by the fine grain size of the products, making microprobe analysis very difficult.

Further difficulties were encountered as a result of the preparation of acceptable starting materials. Oxide mixtures proved less reactive and less homogeneous than nitrate + carbonate mixtures, hence relict unreacted corundum, for example, might be found amongst the products. However, whilst nitrate + carbonate starting materials were more reactive and very homogeneous, they sometimes contained traces of residual nitrates, which acted as fluxes during high-pressure experiments and catalysed melting and breakdown reactions at lowered temperatures. This handicapped the systematization of phase assemblages as a function of temperature.

2. Characterization of  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$

Microprobe analyses of CTA phases in iron-free SYNROC-C formulations showed a narrow composition range, essentially  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ .

An attempt to prepare the pure phase from an oxide mixture of this composition was successful after several iterations of grinding and repelletizing at 1300°C. Powder diffraction X-ray data were obtained for this new phase and are listed in Table 2. Clearly,  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$  is not isostructural phases such as hibonite  $\text{CaAl}_{12}\text{O}_{19}$ , which possess the magnetoplumbite structure.

### 3. Partition of radwaste elements into CTA

The partition behaviour of key elements into CTA has been elucidated by electronprobe analysis. The study included both SYNROC-C formulations and simple two-phase assemblages involving CTA and individual SYNROC phases. Neither  $\text{BaO}$  nor  $\text{ZrO}_2$  show any particular preference for  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ , abundances of 0.2% being typical.  $\text{UO}_2$  and  $\text{Cs}_2\text{O}$  are normally below detection limits. Nickel shows a marked preference for hollandite rather than  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ , rarely exceeding 0.5%  $\text{NiO}$ . Ferric iron is able to substitute readily for  $\text{Al}^{3+}$ , but its solubility limits have not been explored. In perovskite-free SYNROC formulations, several percent of  $\text{SrO}$  is readily accepted by this new phase, thus providing the option of its substituting for perovskite as a host for Sr in SYNROC assemblages. An isostructural phase corresponding to  $\text{Ca}_{.5}\text{Sr}_{.5}\text{Ti}_3\text{Al}_8\text{O}_{19}$  (i.e. 7%  $\text{SrO}$ ) was successfully synthesised at 1320°C and 5 kbar. Small percentages of rare earths have been detected in  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ , however, partition experiments show that perovskite and zirconolite remain the major REE hosts.

### 4. Leaching of Sr-bearing CTA

At 200°C, in pure water,  $\text{Ca}_{.5}\text{Sr}_{.5}\text{Ti}_3\text{Al}_8\text{O}_{19}$  proved very similar to perovskite in terms of Sr leachability (Oversby and Ringwood, 1980). It follows that the capacity of SYNROC to immobilize strontium would be essentially unchanged, if  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$  were to replace perovskite in specific SYNROC-C formulations.

### 5. Stability relations of $\text{CaTi}_3\text{Al}_8\text{O}_{19}$

#### (a) Melting behaviour

Experiments at 5 kbar established that at 1450°C,  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$  melted incongruently producing phenocrysts of a new

Table 1.

CTA phase (mean of 8 analyses)

TiO <sub>2</sub>	34.1
Al <sub>2</sub> O <sub>3</sub>	57.7
CaO	<u>8.2</u>
Sum	<u>100.0</u>

Molecular proportions based on 19 oxygens

Ti	3.005
Al	7.976
Ca	<u>1.026</u>
Sum	<u>12.007</u>

Structural formula     CaTi<sub>3</sub>Al<sub>8</sub>O<sub>19</sub>

Starting composition: TiO<sub>2</sub> = 35.9, Al<sub>2</sub>O<sub>3</sub> = 55.7, CaO = 8.5

Experimental conditions: 5 kbar, 1350°C, 1 hr, Pt caps.

Product assemblage: Texture suggests melting. Euhedral megacrysts of CTA in fine groundmass of perovskite + corundum.

Table 2. Characteristic of spacings ( $\overset{\circ}{\text{\AA}}$ ) of  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$

$\overset{\circ}{\text{d\AA}}$	I
5.6	w
3.79	wm
3.123	w
2.845	s
2.777	w
2.739	w
2.649	wm(b)
2.555	w
2.499	w
2.441	w
2.383	ms
2.267	w
2.182	wm
2.145	m
2.092	ms
1.984	w
1.950	w
1.894	m
1.757	wm
1.738	w
1.636	w
1.609	w
1.601	w
1.528	m
1.509	w(b)
1.436	wm
1.420	m
1.402	ms
1.383	mb

s = strong  
 m = medium  
 w = weak  
 b = broad

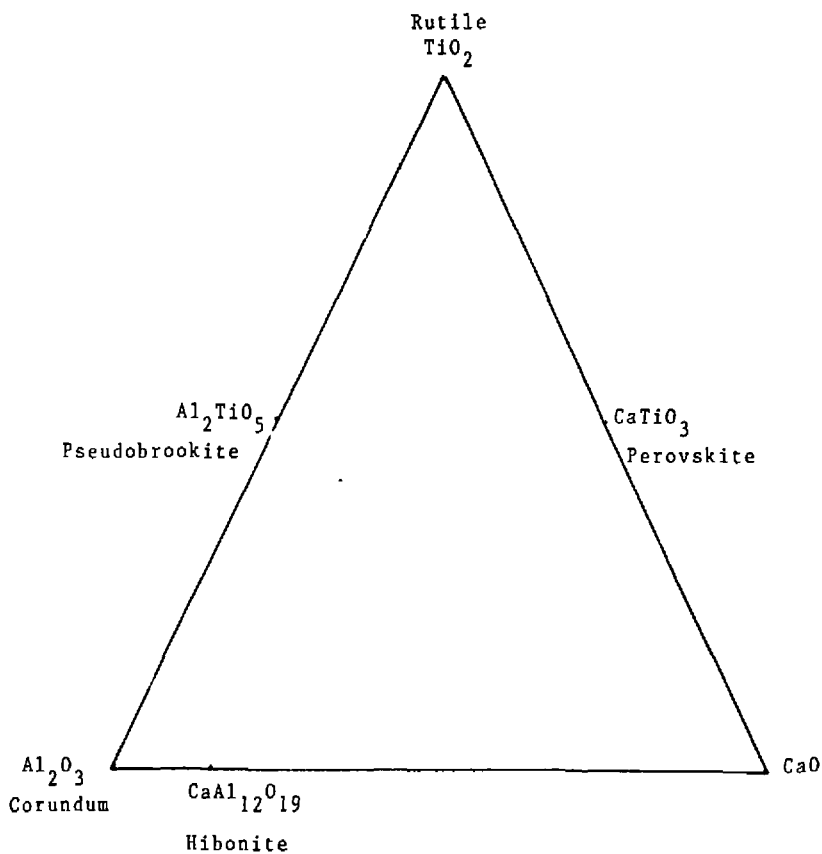
uncharacterized CTA phase (essentially  $\text{Ca}_4\text{Al}_2\text{Ti}_8\text{O}_{23}$ ) in a fine matrix of corundum, perovskite and rutile. The liquidus was not determined accurately but is believed to lie around  $1500^\circ\text{C}$ .

(b) Phase compatibilities in the system  $\text{Al}_2\text{O}_3$ - $\text{TiO}_2$ - $\text{CaO}$

Our experimental study of this system was confined to the area bounded by corundum ( $\text{Al}_2\text{O}_3$ ), rutile ( $\text{TiO}_2$ ) and perovskite ( $\text{CaTiO}_3$ ). Pseudobrookite ( $\text{Al}_2\text{TiO}_5$ ) and CTA ( $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ ) are also contained in this field (Fig. 1). Table 3 summarised the experiments which define the phase relationships and establish the progressive sequence of reactions as a function of increasing temperature. As mentioned in the introduction, certain difficulties beset the interpretation of experimental results.

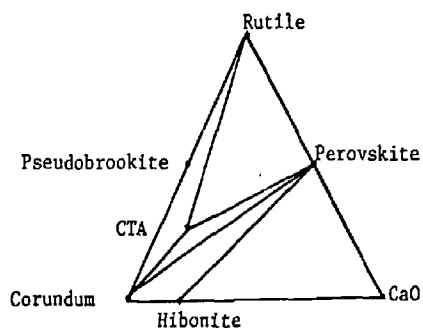
The first eight experiments establish the 3-phase field CTA-perovskite-corundum over a temperature range from 1350 to  $1500^\circ\text{C}$ . Note that at  $1500^\circ\text{C}$ , CTA is replaced by its incongruent melting product (Exp. 5). Experiments 9-11 establish the coexistence of CTA, pseudobrookite and corundum. Despite incomplete reaction in #12, the product assemblage is highly significant because it establishes that under subsolidus conditions, the join CTA-rutile is stable. Subsolidus compatibility relations in the system are illustrated in Figure 2a. Note that perovskite and pseudobrookite are not permitted to coexist at temperatures around  $1300^\circ\text{C}$ . Experiments 13 and 14 show that the join CTA-rutile has been broken, and offer two possible interpretations. The first case is shown in Figure 2b, where a condensed-phase reaction has taken place just below the solidus, with CTA and rutile reacting to yield perovskite + pseudobrookite. The CTA-rutile join disappears and perovskite-pseudobrookite are permitted to coexist. An alternative scenario is presented in Figure 2c. Here, the solidus has been intersected and the small liquid field breaks the CTA-rutile join. The presence of the liquid field also prohibits the coexistence of perovskite and pseudobrookite. (This latter interpretation is preferred, because in SYNROC formulations, perovskite and pseudobrookite are virtually never observed together). The join between CTA and perovskite is probably also cut by the liquid field at these temperatures. The ceramics literature shows that the liquid field breaks the perovskite-rutile join at  $\sim 1460^\circ\text{C}$  and our own experiments, which show that CTA has begun to melt by  $1450^\circ\text{C}$ ,

Fig. 1. Phases in the system  $\text{Al}_2\text{O}_3$ - $\text{TiO}_2$ - $\text{CaO}$

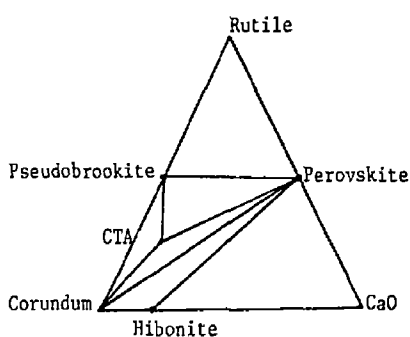


The following phases are not plotted on the diagram:

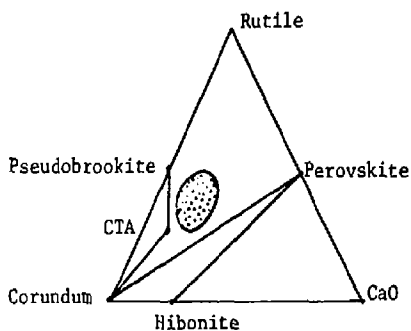
- $3\text{CaO} \cdot 2\text{TiO}_2$
- $4\text{CaO} \cdot 3\text{TiO}_2$
- $3\text{CaO} \cdot \text{Al}_2\text{O}_3$
- $12\text{CaO} \cdot 2\text{Al}_2\text{O}_3$
- $5\text{CaO} \cdot 3\text{Al}_2\text{O}_3$
- $\text{CaO} \cdot \text{Al}_2\text{O}_3$
- $\text{CaO} \cdot 2\text{Al}_2\text{O}_3$



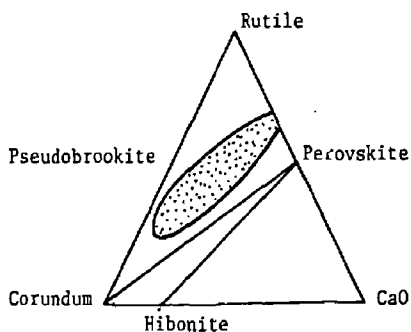
(a) Subsolidus compatibility fields  $\sim 1300^\circ\text{C}$ .



EITHER (b) CTA + rutile + perovskite + pseudobrookite



OR (c) Melting begins  $1300\text{--}1400^\circ\text{C}$ .



(d) By  $\sim 1500^\circ\text{C}$  liquid field eliminates CTA (and any incongruent melting products) breaks rutile-perovskite join.

(e) By  $\sim 1900^\circ\text{C}$ , liquid field eliminates pseudobrookite, perovskite rutile and hibonite.

Fig. 2. Progressive reactions in the system  $\text{Al}_2\text{O}_3\text{--TiO}_2\text{--CaO}$  with increasing temperature.



Table 3. Phase assemblages produced in the system  $\text{Al}_2\text{O}_3\text{-TiO}_2\text{-CaO}$   
All experiments in Pt capsules 0.5-1.0 hrs, 5 kbar.

Exp.	Starting Comp.	Temp. °C	Products	Melted
1.	$\text{CaTi}_4\text{Al}_{10}\text{O}_{24}$ m.o.	1400	CTA, Pvsk, Cor	Yes
2.	" "	1450	" " "	"
3.	" n.c.	1350	" " "	"
4.	$\text{CaTi}_3\text{Al}_8\text{O}_{19}$ n.c.	1400	" " "	No
5.	" "	1500	High-Ca CTA, Pvsk, Cor	Yes
6.	$\text{CaTiO}_3 + \text{Al}_2\text{O}_3$ m.o.	1400	Pvsk, Cor	No
7.	" " "	1500	" "	"
8.	" " n.c.	1500	" "	"
9.	$\text{CaTi}_4\text{Al}_{10}\text{O}_{24}$ m.o.	1400	CTA, Pb, Cor (?Pvsk)	Yes
10.	$\text{CaTi}_3\text{Al}_8\text{O}_{19}$ n.	1400	CTA, Pb, Cor	"
11.	" "	1500	" " "	"
12.	$\text{CaTiO}_3 + \text{Al}_2\text{TiO}_5$ n.c. (1:1 wt)	1300	Ru, mixed Ca-Ti-Al oxides	No
13.	" " m.o.	1400	Pvsk, Pb	Yes
14.	$\text{CaTi}_3\text{Al}_8\text{O}_{19} + \text{TiO}_2$ (1:1 wt) } n.	1400	Pvsk, Pb, Ru	"
15.	$\text{CaTi}_3\text{Al}_8\text{O}_{19} + \text{TiO}_2$ (1:1 wt) } n.c.	1500	Cor, Pvsk, Ru	"

m.o. = starting composition produced by intimately mixing anatase,  $\text{Al}(\text{OH})_3$  and  $\text{CaCO}_3$ .

n.c. = starting composition produced by intimately mixing anatase,  $\text{Al}(\text{NO}_3)_3 \cdot x\text{H}_2\text{O}$  and  $\text{CaCO}_3$ .

n. = starting composition produced by intimately mixing anatase,  $\text{Al}(\text{NO}_3)_3 \cdot x\text{H}_2\text{O}$  and  $\text{Ca}(\text{NO}_3)_2 \cdot x\text{H}_2\text{O}$ .

Pb = pseudobrookite  $\text{Al}_2\text{TiO}_5$

Cor = corundum  $\text{Al}_2\text{O}_3$

Pvsk = perovskite  $\text{CaTiO}_3$

Ru = rutile  $\text{TiO}_2$

CTA =  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$

Table 4. Compatibility between Ba-hollandite ( $\text{BaAl}_2\text{Ti}_6\text{O}_{16}$ ) and  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ .

Starting Composition	Temp. °C	Products	Melted
CTA + Holl (1:1 wt) n.c.	1300	Holl, Ru, fine-grained CTA phases.	No
" " " " "	1350	Holl, Ru, CTA	Yes
" " " m.o.	1300	Holl, Ru, CTA, Cor	No
CTA + $\text{CaZrTi}_2\text{O}_7$ (1:1 wt) m.o.	1350	CTA + Zirc	?

m.o. = starting composition produced by intimately mixing anatase,  $\text{BaCO}_3$ ,  $\text{Al}(\text{OH})_3$  and  $\text{CaCO}_3$ .

n.c. = starting composition produced by intimately mixing anatase,  $\text{BaCO}_3$ ,  $\text{Al}(\text{NO}_3)_3 \cdot x\text{H}_2\text{O}$  and  $\text{CaCO}_3$ .

n. = starting composition produced by intimately mixing anatase,  $\text{Ba}(\text{NO}_3)_2$ ,  $\text{Al}(\text{NO}_3)_3 \cdot x\text{H}_2\text{O}$  and  $\text{Ca}(\text{NO}_3)_2 \cdot x\text{H}_2\text{O}$ .

Holl = hollandite

Ru = rutile

CTA =  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$

Zirc = zirconolite

Cor = corundum

All experiments at 5 kbar, 0.5-1.0 hrs duration, Pt capsules.

Table 5. Magnetoplumbite-type phases produced in SYNROC-C formulations.

	A	B	C	D	E
TiO <sub>2</sub>	12.9	15.4	16.9	11.6	18.0
Al <sub>2</sub> O <sub>3</sub>	72.8	67.6	66.1	72.4	61.5
FeO	8.1	9.6	10.5	8.5	13.8
CaO	5.2	6.4	6.4	6.9	7.4
Sum	98.1	99.0	99.5	99.4	100.7

Structural formulae calculated on the basis of 19 oxygens.

Ti	1.007	1.397	1.529	1.036	1.656
Al	10.247	9.608	9.373	10.137	8.870
Fe	0.809	0.968	1.057	0.844	1.412
Ca	0.665	0.827	0.825	0.878	0.970
Sum	12.80	12.80	12.78	12.90	12.91

Ti+Al+Fe	12.1	12.0	12.0	12.0	11.9
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Experimental conditions: 5 kbar, 1200-1300°C, 1 hr, Fe capsules.

Starting composition: A, B and C = SYNROC + waste, plus 2-3% Fe metal.

D = synthetic magnetoplumbite-type phase.

E = 30% TiO<sub>2</sub>, 47% Al<sub>2</sub>O<sub>3</sub>, 10% FeO, 12% CaO (see text).

place further definition on the size of the liquid field. Thus by about 1500°C the liquid field has expanded to engulf CTA, and joins from CTA to perovskite, corundum and pseudobrookite disappear, as shown in Figure 2d. (For clarity, the stability field of the incongruent melting product of CTA has not been included). Ultimately the liquid field will expand to eliminate pseudobrookite, hibonite, perovskite and rutile.

The phase relations of pure hibonite ( $\text{CaAl}_2\text{O}_6$ ) were not explored in this experimental study. The role of Fe,Ti-bearing magnetoplumbite-type phases in SYNROC formulations is discussed in Section 7 of this paper.

#### 6. Compatibility between CTA and SYNROC minerals

The common occurrence of CTA in many SYNROC-C formulations strongly suggests it is compatible with zirconolite, perovskite and hollandite. The experiments listed in Table 3 establish the coexistence of CTA and perovskite, whilst data in Table 4 confirm that CTA is compatible with hollandite and zirconolite. Its appearance in SYNROC assemblages is generally a result of  $\text{Al}_2\text{O}_3$  activity being higher than normal. This can occur in various ways, for example, entry of  $\text{Ti}^{3+}$  into hollandite "releases"  $\text{Al}_2\text{O}_3$ , which in turn reacts with  $\text{CaO}$  and  $\text{TiO}_2$  in zirconolite and perovskite to produce CTA. CTA has also been produced in SYNROC formulations when the quantity of "excess  $\text{Al}_2\text{O}_3$ " incorporated with the inert additives was greater than that currently recommended (~1%).

#### 7. Characterization of Magnetoplumbite-type phases

Iron-bearing Ca-Ti-Al phases possessing the magnetoplumbite structure have been synthesised in various SYNROC formulations where metallic iron (or cobalt) had been introduced in order to control redox state. Typical examples are given in Cols. A, B, and C of Table 5. Col. D contains the microprobe analysis of one such magnetoplumbite-type phase synthesised separately for X-ray characterization.

An unusual Ca-rich example would contain 30%  $\text{TiO}_2$ , 47%  $\text{Al}_2\text{O}_3$ , 10%  $\text{FeO}$  and 12%  $\text{CaO}$ . It is probably metastable because subsequent attempts to synthesise it alone for further characterization were unsuccessful. The product assemblage instead contained a magnetoplumbite-type phase (Col. 6, Table 5) plus perovskite and Fe-bearing pseudobrookite.

## 8. Crystal chemistry of magnetoplumbite-type phases

The mineral magnetoplumbite ( $\text{PbFe}_{12}\text{O}_{19}$ ) is the type-example of a particular crystal structure  $\text{XY}_{12}\text{O}_{19}$  related to  $\beta\text{-Al}_2\text{O}_3$ . The latter structure is a fast ionic conductor in which large cations e.g. sodium can move freely along conduction planes between spinel-type blocks. In magnetoplumbite, the inter-spinel layers also have unusual 5-fold coordinated sites which can accommodate smaller cations. Morgan et al. (1981) are exploring the suitability of magnetoplumbite phases as components of a ceramic wasteform. They have shown that the "X" cation site may be occupied by Na, Cs, Ca, Sr, Ba,  $\text{Na}_{0.5}$ ,  $\text{REE}_{0.5}$ . Small cations may substitute into the Al, Fe spinel blocks if required for charge-balance. For example  $\text{Ba}^{2+}$  and  $\text{Al}^{3+}$  in  $\text{BaAl}_{12}\text{O}_{19}$  can be replaced by  $\text{La}^{3+}$  and  $\text{Mg}^{2+}$  to give a new magnetoplumbite-type phase,  $\text{LaMgAl}_{11}\text{O}_{19}$ .

Keil and Fuchs (1971) have studied the mineral hibonite, a calcium-aluminate magnetoplumbite-type phase which occurs naturally in meteorites, and have found that Cr, Ti and Mg may enter the structure. We have successfully synthesised a magnetoplumbite-type phase with half its  $\text{Ca}^{2+}$  replaced by  $\text{Sr}^{2+}$ . However, in SYNROC assemblages, perovskite and not the magnetoplumbite-type phase remains the major host for strontium. Our exploratory studies of a high-alumina SYNROC-D wasteform (Kesson and Ringwood, 1981) revealed that Mn and Na could also be accommodated, but there is as yet no independent evidence to confirm the belief of Morgan et al. (1981) that both Cs and Si can simultaneously enter the magnetoplumbite structure.

The analyses of magnetoplumbite-type phases in Table 5 are characterised by deficient cation sums for a structure based on 19 oxygen atoms. The structural formulae reveal that the deficiency is related to vacancies in large-cation ("X") sites, whilst the "Y" sites, occupied by Al, Ti and Fe, remain essentially full. Interestingly, near-integral cation sums ( $\sim 2$ ) are obtained in most cases for structural formulae based on 3 oxygens. The significance of this is not understood.

## 9. Compatibility relations of magnetoplumbite-type phases in SYNROC-C

Magnetoplumbite-type phases coexists with hollandite, perovskite, zirconolite and Fe-bearing pseudobrookite. They do not

coexist with the CTA phase  $\text{CaTi}_3\text{Al}_8\text{O}_{19}$ . (In the simple system  $\text{Al}_2\text{O}_3$ - $\text{TiO}_2$ - $\text{CaO}$  the corundum-perovskite join prohibits CTA from coexisting with hibonite  $\text{CaAl}_{12}\text{O}_{19}$ , a magnetoplumbite-type structure).

In SYNROC formulations contaminated with iron, magnetoplumbite-type phases may appear together with hollandite, zirconolite and perovskite. Note that the magnetoplumbite-type phases alone is observed, never an Fe-bearing form of the particular CTA phase discussed in Section 2 of this manuscript. Because the manufacture of SYNROC will be carried out under conditions where oxygen fugacities are controlled at low levels, we have not explored the effects of  $\text{Fe}^{3+}$  in stabilizing CTA versus magnetoplumbite-type phases.

#### 10. Elimination of magnetoplumbite-type phases from SYNROC-C formulations

Magnetoplumbite-type phases are structurally related to the fast ionic conductor  $\beta\text{-Al}_2\text{O}_3$ , and so we suspect its resistance to groundwater leaching may be far from optimum. It would therefore seem preferable to eliminate it from SYNROC assemblages. Its appearance is related to FeO contamination which occurs when metallic iron is introduced into SYNROC precursors in order to control oxygen fugacities at low levels during hot-pressing. The metallic iron inevitably reduces some  $\text{Ti}^{4+}$  to  $\text{Ti}^{3+}$  and is consequently oxidized to FeO. This problem can be eliminated very easily by substituting metallic titanium for metallic iron, as the redox control agent.

#### 11. References

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# Immobilization of Highly Aluminous Sludges

S.E. Kesson and A.E. Ringwood

## 1. Introduction

Two new SYNROC formulations aimed at immobilizing very alumina-rich sludges have been explored experimentally. In the first of these, magnesia ( $MgO$ ) is included with the other inert SYNROC additives ( $TiO_2$ ,  $ZrO_2$ ,  $SiO_2$ ,  $CaO$ , etc.) with the aim of producing a wasteform comprised of zirconolite, perovskite, nepheline and Mg-spinel solid solution. This treatment is analogous to the manufacture of regular SYNROC-D, where iron serves to immobilize alumina as  $FeAl_2O_4$  component in spinel. The second alternative involves adding a minimum of inert oxide additives, with the aim of allowing "surplus" alumina in the sludges to recrystallize as corundum during wasteform manufacture.

## 2. Sludge Composition

The composition of the high-Al sludge is listed in Column 1, Table 1. After calcining under controlled redox conditions, the sludge composition should be approximately that given in Column 2, Table 1. This latter composition was used as the starting point for our experiments. Note that zepolites have been excluded from the waste stream and various oxide species recast to others more appropriate to SYNROC redox conditions, e.g.  $U_3O_8 + UO_2$ ,  $MnO_2 + MnO$ ,  $Fe_2O_3 + Fe_3O_4$ .

## 3. SYNROC-D; Mg-formulation

### (a) Experimental strategy

In this instance, it is desirable that the final product should consist of zirconolite, perovskite, MgO-bearing spinel ss. and nepheline, and, moreover, there should be a minimum of 10 wt.% each zirconolite and perovskite, in order to allow for "flexibility" in handling potential waste stream variations. The appropriate proportions of inert additives (including MgO) have been calculated accordingly. The results of this exercise are presented in Table 2, which shows that 46.5% of inert additives would be required to immobilize 53.5% calcined sludge. Ideally, the final SYNROC-D



Table 1. High-Al sludge compositions.

<u>High-Al Sludge</u>		<u>Calcine</u>	
SiO <sub>2</sub>	0.6	SiO <sub>2</sub>	0.6
MnO <sub>2</sub>	4.4	UO <sub>2</sub>	1.4
U <sub>3</sub> O <sub>8</sub>	1.3	Al <sub>2</sub> O <sub>3</sub>	84.8
Al <sub>2</sub> O <sub>3</sub>	76.1	Fe <sub>3</sub> O <sub>4</sub>	5.7
Fe <sub>2</sub> O <sub>3</sub>	5.3	MnO	4.0
CaO	0.4	CaO	0.4
NiO	0.8	NiO	0.9
Na <sub>2</sub> O	2.0	Na <sub>2</sub> O	2.2
SO <sub>4</sub> <sup>2-</sup>	<0.5	Sum	100.0
zeolite	9.3		
Sum	100.7		

The composition of the calcine (Col. 2) has been derived from the composition of the sludge (Col. 1) in the following manner.

- (i) Zeolite has been excluded.
- (ii) Certain oxide species have been converted to others more appropriate to hot-pressing redox conditions
  - $\text{Fe}_2\text{O}_3 \rightarrow \text{Fe}_3\text{O}_4$  (enters spinel ss. as  $\text{Fe}^{2+}$  and  $\text{Fe}^{3+}$ )
  - $\text{MnO}_2 \rightarrow \text{MnO}$  (enters spinel ss. as  $\text{Mn}^{2+}$ )
  - $\text{U}_3\text{O}_8 \rightarrow \text{UO}_2$  (enters zirconolite as  $\text{U}^{4+}$ )
- (iii) Sulfate will be lost during calcining.
- (iv) Sum recalculated to 100.

wasteform would consist of 6 wt.% nepheline, 12 wt.% each zirconolite and perovskite, and 70 wt.% spinel s.s.

Starting material corresponding to the bulk composition given in the final column of Table 2 was prepared from an intimate well-ground mixture of oxides, hydroxides and carbonates, following our normal laboratory procedures. It was then hot-pressed at 5 kbar and temperatures of 1150°C, 1200°C and 1250°C for 1 hr. Platinum sample capsules were used, so that oxygen fugacity conditions near Ni-NiO would prevail during hot-pressing.

(b) Experimental results

X-ray diffraction studies showed that at 1100°C, perovskite and spinel were well-crystallized. The characteristic zirconolite lines could also be identified. In the higher temperature runs, the intensities of the zirconolite lines increased relative to perovskite and spinel. The characteristic lines of nepheline could not be confirmed in any run, however, with only ~6% nepheline in the assemblage this is not entirely unexpected. Relict  $\text{TiO}_2$  was also identified in the 1100°C run. In polished section, the 1100°C and 1200°C runs were very fine-grained, whilst the poor polish on the 1100°C strongly suggested incomplete reaction. This was subsequently confirmed by microprobe examination, which revealed traces of relict  $\text{TiO}_2$  and  $\text{ZrO}_2$ . The 1250°C run consisted of a homogeneous equigranular material, with grain size 1-10 $\mu$ , with spinel typically forming the larger grains. Representative microprobe analyses of zirconolite and spinel are presented in Table 3. Perovskite was too finely crystalline for successful analysis - and the presence of nepheline was never confirmed with certainty in any run, probably due to its fine grain size and low abundance.

(c) Discussion

In principle, highly aluminous sludges could be immobilized using special SYNROC formulations which include MgO amongst the inert additives. The product wasteform would consist of zirconolite, perovskite, nepheline and spinel ss., with the MgO serving to immobilize the bulk of the alumina as  $\text{MgAl}_2\text{O}_4$  component in spinel solid solution. One advantage of this process would be the comparative ease with which redox conditions could be controlled. However, there are some potential problems with the strategy.

Table 2. SYNROC-D; Mg-formulation.

	<u>53.5% SLUDGE CALCINE</u>	<u>46.5% INERT ADDITIVES</u>	<u>BULK SYNROC</u>
SiO <sub>2</sub>	0.3	2.0	2.3
TiO <sub>2</sub>	-	15.2	15.2
ZrO <sub>2</sub>	-	3.9	3.9
UO <sub>2</sub>	0.8	-	0.8
Al <sub>2</sub> O <sub>3</sub>	45.4	-	45.4
Fe <sub>3</sub> O <sub>4</sub>	3.1	-	3.1
MgO	-	18.8	18.8
MnO	2.1	-	2.1
NiO	0.5	-	0.5
CaO	0.2	6.3	6.8
Na <sub>2</sub> O	1.2	-	1.2
Sum	53.6	46.5	100.1

Ideal Mineralogy (wt.%)    Nepheline ....6  
                                  Zirconolite...12  
                                  Perovskite....12  
                                  Spinel s.s....70

Table 3. Analyses of spinel and zirconolite in Mg-bearing SYNROC-D.

	<u>Spinel ss.</u>	<u>Zirconolite</u>
TiO <sub>2</sub>	4.8	42.9
ZrO <sub>2</sub>	-	31.8
UO <sub>2</sub>	-	5.5
Al <sub>2</sub> O <sub>3</sub>	62.2	1.1
MgO	26.2	0.7
FeO	3.8	1.3
MnO	2.8	0.9
NiO	0.3	-
CaO	0.3	14.4
Sum	<u>100.4</u>	<u>98.6</u>

Perovskite and nepheline too fine-grained for analysis.

Firstly, it appears that equilibrium is achieved only over a narrow temperature interval near the solidus. Moreover, the temperatures involved are quite high (around 1200°C) and careful control of hot-pressing conditions would be critical. Other problems would arise if the proposal to incorporate fragmented Cs-bearing hollandite into SYNROC-D were followed. At 1200°C, the hollandite melts and reacts vigorously with the SYNROC matrix, and no longer serves to immobilize cesium.

#### 4. SYNROC-D : Corundum formulation

##### (a) Experimental strategy

In this formulation, the aim is to incorporate the minimum amount of inert additives to yield a product assemblage consisting of zirconolite and perovskite (a minimum of 10 wt.% of each) plus nepheline, spinel ss. and corundum. The mineralogy of the product wasteform would differ significantly from other SYNROC-D formulations in that the modal proportion of spinel would be relatively small. The spinel phase is produced by the combination of iron, manganese and nickel in the sludges with some alumina. The "excess" alumina remaining should crystallize as abundant corundum. The product assemblage might also contain other aluminous phases, such as a magnetoplumbite-type phase ( $XY_{12}O_{19}$ ), the CTA phase  $CaTi_3Al_8O_{19}$ , or pseudobrookite. (This latter phase would necessarily be fairly Fe-rich, in order to coexist with perovskite since pure  $Al_2TiO_5$  and perovskite are incompatible). The experimental strategy was aimed at determining whether the above strategy were viable, and at confirming the mineralogy of the product assemblage.

The proportions of inert additives (Table 4) were calculated according to the requirements that all sodium in the sludges be converted to nepheline, and that there should be about 10 wt.% each of perovskite and zirconolite in the final product. The resultant waste-loading is 74%.

Starting material corresponding to the bulk composition in Column 3, Table 4 was prepared from mixed oxides, hydroxides and carbonates according to our preferred procedures. Hot-pressing was carried out for 1 hr. at temperatures of 1200°C in Ni and Pt capsules, and at 1250°C and 1300°C in Ni capsules. The products were examined by X-ray diffraction and electron microprobe analysis.

Table 4. SYNROC-D : Corundum formulation.

	<u>74% SLUDGE CALCINE</u>	<u>26% INERT ADDITIVES</u>	<u>BULK SYNROC</u>
SiO <sub>2</sub>	0.4	2.7	3.1
TiO <sub>2</sub>	-	12.9	12.9
ZrO <sub>2</sub>	-	4.0	4.0
UO <sub>2</sub>	1.0	-	1.0
Al <sub>2</sub> O <sub>3</sub>	62.7	-	62.7
Fe <sub>3</sub> O <sub>4</sub>	4.2	-	4.2
MnO	3.0	-	3.0
NiO	0.7	-	0.7
CaO	0.3	6.4	6.7
Na <sub>2</sub> O	1.6	-	1.6
Sum	<u>73.9</u>	<u>26.0</u>	<u>99.9</u>

Ideal mineralogy (wt.%)	Nepheline..... 7
	Zirconolite..... 9
	Perovskite .... 12
	Spinel ss..... 15
	Corundum..... 57

Table 5. Magnetoplumbite-type phases crystallized from SYNROC-D (Corundum formulation).

	A	B	C	D	E
TiO <sub>2</sub>	10.1	11.4	10.5	10.3	14.1
Al <sub>2</sub> O <sub>3</sub>	73.3	72.2	72.7	73.8	68.1
FeO	3.5	4.1	3.4	4.1	3.7
NiO	0.8	0.2	2.4	0.0	1.4
MnO	3.4	3.6	3.2	3.1	3.4
CaO	6.8	6.6	6.5	6.8	7.9
Na <sub>2</sub> O	1.3	1.0	1.2	1.5	0.8
Sum	99.2	99.1	99.9	99.6	99.4

Structural formulae based on 19 oxygens

Ti	0.903	1.021	0.938	0.916	1.276
Al	10.274	10.139	10.179	10.280	9.655
Fe	0.348	0.409	0.338	0.405	0.372
Ni	0.343	0.019	0.229	0.0	0.136
Mn	0.077	0.363	0.322	0.310	0.346
Ca	0.866	0.843	0.827	0.861	1.018
Na	0.300	0.231	0.276	0.344	0.187
Sum	13.11	13.02	13.11	13.12	12.99

Starting composition: "Bulk SYNROC" as in Table 4.

Experimental conditions: 5 kbar, 1 hr, A at 1200°C Ni capsule  
 B " 1200°C Pt "  
 C " 1250°C Ni "  
 D " 1100°C Ni "  
 E " 1300°C Ni "

### (b) Experimental results

The product assemblages consisted of abundant needles ( $\sim 20 \mu$ ) of a magnetoplumbite-type phase in a fine-grained matrix. Neither spinel, nepheline, corundum nor perovskite were identified by microprobe or X-ray, although minor zirconolite and baddelyite were observed. Analysis of the magnetoplumbite-type phases (Table 5) revealed that they were similar to hibonite ( $\text{CaAl}_{12}\text{O}_{19}$ ) with minor substitution of Na for Ca and Mn, Ti, Ni and Fe for Al.

The fine grain size of the matrix in the above experiments did not permit microprobe analysis of phases coexisting with the magnetoplumbite. A second series of experiments on a moderately aluminous sludge composition carried out just above the solidus temperature established that perovskite, zirconolite, spinel, a Cs-rich "nepheline" and Cs-free nepheline could coexist with the magnetoplumbite-type phase. An example of this compatible phase assemblage is presented in Table 6.

### (c) Discussion

The various experiments described above demonstrate that it is not possible to convert highly-aluminous sludges to a wasteform comprising zirconolite, perovskite, spinel and nepheline plus corundum, by adding appropriate minimum amounts of the inert SYNROC additives ( $\text{CaO}$ ,  $\text{TiO}_2$ ,  $\text{ZrO}_2$ ,  $\text{SiO}_2$ ). The product mineralogy is instead dominated by a calcium-aluminate phase with the magnetoplumbite structure. This phase was found to accommodate minor amounts of Ti, Mn, Ni, Fe and Na in solid solution.

Scientists working for Rockwell International have advocated that highly-aluminous sludges be converted to a wasteform whose mineralogy is dominated by a magnetoplumbite-type phase (e.g. Morgan et al., 1981; Harker et al., 1981). A very similar product has been obtained in the above series of experiments as a direct result of following the SYNROC strategy first described by Ringwood (1978) and subsequently modified for defense waste immobilization (Ringwood et al., 1980).

## 5. Conclusion

Geological data lead us to suspect that a wasteform dominated by the inert and refractory phase corundum would have been superior to one containing a magnetoplumbite-type phase, in terms of its resistance to groundwater leaching. For this reason we do



Table 6. Magnetoplumbite-type phase plus coexisting phases in moderately-aluminous SYNROC-D formulation.

	Bulk	Zirc	Pvsk	Sp	Cs-neph	Mtpb
SiO <sub>2</sub>	6.8	-	-	-	33.7	-
TiO <sub>2</sub>	17.5	34.6	53.6	3.9	-	15.1
ZrO <sub>2</sub>	5.5	33.6	0.3	-	-	-
UO <sub>2</sub>	2.0	10.5	3.9	-	-	-
Al <sub>2</sub> O <sub>3</sub>	20.1	1.8	1.3	49.5	28.3	61.2
REE*	3.	<.5	1.5	-	-	-
FeO	22.8	4.8	2.3	36.1	0.5	14.5
MnO	5.3	1.7	0.7	9.3	0.3	2.7
NiO	2.5	-	tr.	1.0	-	1.4
CaO	7.1	11.3	34.1	-	0.4	4.8
Na <sub>2</sub> O	4.0	0.3	0.2	-	11.6	-
Cs <sub>2</sub> O	0.3	SrO=1.1	SrO=1.0	-	26.0	-
Sum	100.4	99.7	98.9	99.8	100.8	99.7

Also present: Glass of variable composition; nepheline, essentially NaAlSiO<sub>4</sub>.

"BULK" = zirconolite  
Pvsk = perovskite  
Sp = spinel ss.  
Cs-neph = Cs-rich nepheline  
Mtpb = magnetoplumbite-type phase

Experimental conditions: 5 kbar, 1200°C, Ni capsule, 1 hr.

\*REE = Ce<sub>2</sub>O<sub>3</sub> + Nd<sub>2</sub>O<sub>3</sub>

not advocate converting highly-aluminous sludges to the latter mineralogy. The strategy described earlier in this paper, whereby the "excess" alumina was immobilized as  $\text{MgAl}_2\text{O}_4$  spinel requires comparatively high temperature in order to achieve an equilibrium product assemblage. However, our preferred approach would be to incorporate FeO rather than MgO, which would similarly serve to immobilize alumina as  $\text{FeAl}_2\text{O}_4$  spinel component but at lower hot-pressing temperatures. This latter option could be readily carried out by mixing high-iron sludges with high-alumina sludges, and eliminating the handling of high-alumina sludges as a separate waste-stream.

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## SYNROC-D FORMULATIONS PRODUCED BY SINTERING IN AIR

### INTRODUCTION

SYNROC-D formulations adapted for the immobilization of the average Savannah River composite sludge (Table 1) have previously been prepared according to recommended procedures, by hot-pressing under controlled redox conditions (reducing) so that a large proportion of iron is present in the divalent state. In this report we investigate the possibility of preparing SYNROC-D under oxidising conditions by the simpler procedure of sintering in air, at temperatures appreciably above the solidus (1200-1250°C). A dense (95%) material with low porosity can be produced in this manner. Its mineralogy and phase chemistry have been investigated, and the partition behaviour of key radwaste elements such as strontium and uranium has been established.

### EXPERIMENTAL PROCEDURES

Pelletized fragments of SYNROC-D were suspended in Pt containers in vertical furnaces for periods of 1-5 hours. The products were examined by X-ray diffraction and electron microprobe.

### PHASE CHEMISTRY OF SYNROC-D AFTER SINTERING IN AIR

The SYNROC-D material produced by sintering in air at 1200-1250°C does not possess the typical phase assemblage (zirconolite, perovskite, spinel and nepheline). Instead, one or more new oxide phases plus a low-silica glass are observed and primary nepheline was not positively identified.

The accompanying Tables 2 and 3 present analyses of coexisting phases produced in two sintering runs at 1200°C. The 1½ hour run (Table 2) yielded a phase assemblage of fine-grained perovskite, rare ilmenite, intergranular films of glass, a hibonite-type phase of variable composition and two chemically-distinct zirconolites, differing mainly in their UO<sub>2</sub> and ZrO<sub>2</sub> contents. A longer run of 5 hours duration (Table 3) yielded fine-grained perovskite, zirconolite, a hibonite-type phase, and nepheline. The unusually high TiO<sub>2</sub>, Fe<sub>2</sub>O<sub>3</sub> and CaO

TABLE 1  
 SYNROC-D FORMULATION DESIGNED TO IMMOBILIZE  
 AVERAGE SAVANNAH RIVER SLUDGE

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$\text{SiO}_2$	7.4
$\text{TiO}_2$	18.8
$\text{ZrO}_2$	5.6
$\text{UO}_2$	2.2
$\text{Al}_2\text{O}_3$	19.8
$\text{Ce}_2\text{O}_3$	0.6
$\text{Nd}_2\text{O}_3$	0.6
$\text{Fe}_2\text{O}_3$	25.3
$\text{MnO}$	5.1
$\text{NiO}$	3.3
$\text{CaO}$	7.3
$\text{SrO}$	0.3*
$\text{Na}_2\text{O}$	3.5
$\text{Cs}_2\text{O}$	0.3
<hr/>	
SUM	100.1

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This composition was used in the SYNROC-D sintering experiments as described in this report.

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\* An additional 1%  $\text{SrO}$  was subsequently added for partition experiments.

TABLE 2  
PHASES PRODUCED IN SYNROC-D AFTER SINTERING  
IN AIR FOR 1½ HOURS AT 1200°C.

	<u>Zirconolite</u>		<u>Hibonite-type phase</u>		<u>Ilmenite</u>
	(a)	(b)	variable composition		
TiO <sub>2</sub>	31.7	26.3	17.9	→ 12.7	38.5
ZrO <sub>2</sub>	32.7	43.9	--	--	--
UO <sub>2</sub>	10.7	5.0	--	--	--
CeO <sub>2</sub>	2.6	1.5	--	--	--
Al <sub>2</sub> O <sub>3</sub>	1.4	2.4	30.7	43.0	7.6
Fe <sub>2</sub> O <sub>3</sub>	7.0	8.7	34.0	28.4	19.4
FeO	--	--	--	--	29.8
NiO	0.3	0.8	3.9	3.1	0.5
MnO	1.4	2.0	7.0	6.0	3.9
CaO	11.9	10.2	6.3	5.1	0.2
SrO	<.2	<.2	--	--	--
Na <sub>2</sub> O	<.2	<.2	1.5	1.5	--
SUM	99.7	100.5	101.2	99.8	99.9

Perovskite too fine-grained for satisfactory analysis.

Glassy Na, Si-rich material at grain boundaries.

Experimental Conditions: "SC" composition sintered in air at  
1200 C, 1½ hours.

All iron expressed as Fe<sub>2</sub>O<sub>3</sub> in zirconolite and hibonite.

TABLE 3  
PHASES PRODUCED IN SYNROC-D AFTER SINTERING  
IN AIR FOR 5 HOURS AT 1200°C

	<u>Zirconolite</u>	Hibonite-type phase	(Quench) Nepheline
SiO <sub>2</sub>	--	--	41.5
TiO <sub>2</sub>	32.1	16.7	0.5
ZrO <sub>2</sub>	32.0	--	--
UO <sub>2</sub>	10.5	--	--
REE*	2.5	--	--
Al <sub>2</sub> O <sub>3</sub>	1.6	26.7	35.8
Fe <sub>2</sub> O <sub>3</sub>	8.6	38.0	1.7
FeO	--	--	--
NiO	0.2	4.6	--
MnO	1.4	7.5	0.1
CaO	12.1	4.6	3.3
SrO	0.6	0.8	--
Na <sub>2</sub> O	<.2	1.7	17.2
SUM	101.4	100.6	100.3

Perovskite is present but way too fine-grained for satisfactory analysis; however, it contained both Nd<sub>2</sub>O<sub>3</sub> and SrO (~4%), and UO<sub>2</sub> (~0.5%).

Glass also present at grain boundaries.

Experimental conditions: "SC" composition sintered in air at 1200°C for 5 hours.

All iron expressed as Fe<sub>2</sub>O<sub>3</sub>.

\* REE = CeO<sub>2</sub> + Nd<sub>2</sub>O<sub>3</sub>

contents of the nepheline are consistent with its formation during quenching. Intergranular glassy material is also present but could not be analysed successfully. A better approach to an equilibrium assemblage was probably obtained in this 5 hour run, as evidenced by the narrow composition range for both zirconolite and the hibonite-type phase, and the absence of ilmenite.

Positive confirmation of the identity of the hibonite-type phase which appears in the above experiments was achieved by means of a separate series of synthesis experiments on a bulk composition similar to that in col. 2, Table 3, whereby pelletized fragments were heated in air at 1200, 1250 and 1300°C for 1-2 hours. At 1200°C reaction was incomplete. At 1250°C and 1300°C the products consisted largely of a hibonite-type phase (col. 1, Table 4) plus trace amounts of another uncharacterized Ca-Ti-Al-Fe phase. The crystal structure of the hibonite-type phase was confirmed from its X-ray diffraction pattern. (Hibonite,  $\text{CaAl}_{12}\text{O}_{19}$ , is isostructural with magneto-plumbite ( $\text{PbFe}_{12}\text{O}_{19}$ ). The structural formula indicates some solid solution towards  $\beta\text{-Al}_2\text{O}_3$  ( $\text{NaAl}_{11}\text{O}_{17}$ ) because the cation sum ( $\text{Ca} + \text{Na}$ ) is slightly in excess of unity.  $\text{Ti}^{4+}$  must enter the structure via a coupled substitution whereby  $\text{Ti}^{4+} + \text{M}^{2+}$  exchange for two trivalent ions ( $\text{Fe}^{3+}$  and  $\text{Al}^{3+}$ ). There is insufficient  $\text{Mn}^{2+}$  and  $\text{Ni}^{2+}$  to balance all the  $\text{Ti}^{4+}$  hence some of the iron must also be present in the divalent state, although this would amount to less than 15% of the total amount of iron in the phase. It follows that most of the iron is present in the +3 state, as might be expected from the highly oxidising synthesis conditions.

It is evident that whilst some SrO enters both hibonite and zirconolite, perovskite remains the major host for Sr. Substantial amounts of  $\text{UO}_2$  are taken up by zirconolite, however some uranium, presumably in the hexavalent state, may occur in the glass. More specific evidence for this was obtained in runs at 1225°C and 1250°C where the glasses commonly contained from 1 to 5% of  $\text{UO}_2$ . Rare earths are partitioned into zirconolite and perovskite and were not detected in glass or hibonite. Most  $\text{Cs}_2\text{O}$  is lost during the heat-treatment, however any residual Cs would be expected to partition into the glass.

TABLE 4  
SYNTHESIS OF HIBONITE-TYPE PHASE

	Hibonite-type phase	Rare uncharacterised Ca-Ti-Al-Fe phase
TiO <sub>2</sub>	14.8	28.2
Al <sub>2</sub> O <sub>3</sub>	34.1	20.5
Fe <sub>2</sub> O <sub>3</sub>	36.2	25.3
MnO	5.7	4.7
NiO	3.2	3.1
CaO	4.6	16.7
Na <sub>2</sub> O	1.5	1.1
SUM	100.1	99.6

Structural formula based on 19 oxygens. Fe<sup>'''</sup> and Fe<sup>''</sup>  
calculated assuming atomic (Mn + Ni + Fe<sup>''</sup>) = Ti

Ti	1.541	
Al	5.566	
Fe <sup>'''</sup>	3.257	
Fe <sup>''</sup>	0.516	
Mn	0.669	} 1.025
Ni	0.356	
Ca	0.683	} 1.083
Na	0.403	
SUM	12.991	

Experimental conditions: sintered in air at 1300°C for 1 hour.



Sintering SYNROC-D in air at 1225°C for 3 hours resulted in a different phase assemblage in which zirconolite was destabilized in favour of a new face-centred cubic (defect-fluorite-type) structure. The product phase assemblage is shown in Table 5.

The identity of the new defect fluorite-type phase was confirmed in separate synthesis experiments involving sintering in air at 1250, 1300 and 1350°C for 0.5 and 2 hours. The products at all temperatures consisted of a two-phase assemblage, predominantly the defect-fluorite-type phase, plus minor zirconolite. An example of the coexisting pair is provided in Table 6. It is difficult to deduce much about the crystal chemistry of this new phase, except to note that a large proportion of its iron is probably in the ferrous state, because a good value for the oxide sum (99.5%) is obtained with all iron expressed as FeO.

The partition behaviour of strontium differs from that observed at 1200°C, with larger amounts entering the hibonite-type phase compared to perovskite, and some also partitioning into the new defect-fluorite-type phases. Zirconolite, as noted earlier, is no longer stable, and so perovskite serves as the main host for rare earths. Although some  $\text{UO}_2$  is contained in the defect-fluorite-type phases, a significant amount can be found in the interstitial glass, where it presumably occurs in the hexavalent state. The leaching performance of a wasteform with so much of its uranium in glass rather than in crystal lattice sites, is likely to be rather poor.

After sintering in air for 1.5 hours at 1250°C further modification of the phase assemblage was observed. The product consisted of small perovskite grains, together with abundant hibonite-type and defect-fluorite-type phases, the latter showing a range of composition. Zirconolite was again absent, however a ferric spinel was observed. (This is the first occurrence of spinel in this particular experimental study.) Interstitial glass was also present. Analyses of coexisting phases are presented in Table 7.

TABLE 5  
PHASES PRODUCED IN SYNROC-D AFTER SINTERING  
IN AIR FOR 3 HOURS AT 1225°C

	Hibonite-type phase	Defect-fluorite-type (a) phases (b)		Glass
SiO <sub>2</sub>	--	--	--	31.5
TiO <sub>2</sub>	16.0	32.2.	31.2	13.7
ZrO <sub>2</sub>	--	22.8	13.9	0.9
UO <sub>2</sub>	--	5.2	1.0	4.8
REE	--	--	1.0	--
Al <sub>2</sub> O <sub>3</sub>	26.2	5.7	9.9	21.9
Fe <sub>2</sub> O <sub>3</sub>	38.8	--	--	--
FeO	--	16.6	25.0	6.5
MnO	7.1	3.7	5.4	2.2
NiO	4.6	0.5	1.2	0.1
CaO	4.6	10.5	9.9	11.2
SrO	3.0	0.3	1.0	--
Na <sub>2</sub> O	1.2	0.3	0.7	6.4
SUM	101.3	99.6	100.2	99.2

Fine-grained Perovskite contains 1.2% SrO, 0.2% UO<sub>2</sub> and  
~5% rare earths.

All iron expressed as Fe<sub>2</sub>O<sub>3</sub> in hibonite, otherwise all as FeO.

TABLE 6  
COEXISTING DEFECT-FLUORITE-TYPE PHASE AND ZIRCONOLITE

	DEFECT-FLUORITE- TYPE PHASE	ZIRCONOLITE
TiO <sub>2</sub>	31.8	31.5
ZrO <sub>2</sub>	14.3	34.2
UO <sub>2</sub>	1.8	6.5
REE	~1.0	~3.0
Al <sub>2</sub> O <sub>3</sub>	10.7	1.5
FeO	22.7	8.7
MnO	5.0	1.4
NiO	1.8	0.4
CaO	9.3	11.8
SrO	0.4	0.9
Na <sub>2</sub> O	0.7	0.3
SUM	99.5	100.2

Experimental conditions: Pellet sintered in air at 1350°C  
for 0.5 hours.

TABLE 7  
COEXISTING PHASES PRODUCED IN SYNROC-D AFTER  
SINTERING IN AIR FOR 1½ HOURS AT 1250°C.

	Hibonite- type phase	Defect-fluorite-type phase (variable composition)		Spinel	Glass
SiO <sub>2</sub>	--	--	--	--	34.2
TiO <sub>2</sub>	15.1	31.7	29.7	2.5	4.0
ZrO <sub>2</sub>	--	26.6	14.4	--	0.2
UO <sub>2</sub>	--	7.4	2.2	--	3.4
REE	--	~1.	~2.	--	<.2
Al <sub>2</sub> O <sub>3</sub>	32.3	4.1	9.7	14.0	29.0
Fe <sub>2</sub> O <sub>3</sub>	37.2	--	--	47.6	6.6
FeO	--	11.6	23.3	3.9	--
MnO	5.3	2.4	5.2	15.6	1.9
NiO	4.3	1.5	2.8	16.3	1.2
CaO	4.8	10.6	8.5	0.2	4.6
SrO	0.4	0.3	0.4	--	--
Na <sub>2</sub> O	1.5	0.3	0.4	--	12.6
SUM	100.9	97.5	98.9	99.9	97.7

Fine-grained perovskite also present.

Experimental conditions: Sintered in air at 1250°C for 1.5 hours.

Because perovskite is too small for satisfactory microprobe analysis it is not possible to confirm the partition behaviour of strontium, although mass-balance consideration suggest that perovskite remains the primary strontium host. Some uranium is contained in the defect-fluorite-type phase, however, significant amounts are also present in the glass, presumably in the hexavalent state, and potentially readily leached. Most of the iron in the spinel and hibonite-type phases is in the oxidised state, whereas the defect-fluorite-type phase appears to contain both ferric and ferrous iron.

### CONCLUSIONS

Sintering of SYNROC-D in air at temperatures just above the solidus (1200-1250°C) produces a dense compact ceramic, however the optimum SYNROC-D mineralogy is not produced. Although perovskite is present throughout, spinel appears only at 1250°C, whilst zirconolite is destabilized in favour of a defect-fluorite-type phase above 1200°C. A new, highly oxidised phase structurally similar to hibonite is common throughout and may contain significant amounts of strontium and sodium. Nepheline was identified only as a byproduct of quenching. The interstitial liquid contains substantial amounts of uranium which may well be readily leachable. Moreover, any neptunium present in the waste would probably be converted to a highly oxidised species which would preferentially partition into the glassy phase. Most cesium is lost by volatilization. A SYNROC-D wasteform produced by sintering in air is not expected to display good leaching behaviour for U, Np and other actinides. For this reason, production of SYNROC-D by air - sintering is not recommended.

The wide field of stability of the hibonite-type phase is interesting. It co-exists with zirconolite, perovskite, spinel and probably with nepheline. It has previously been encountered when hot-pressing SYNROC-D in platinum capsules under moderately oxidising conditions (early report by A.E. Ringwood to J. Tewhey) but its crystal structure was not identified.

The hibonite phase (ideally  $\text{CaAl}_{12}\text{O}_{19}$ ) present in these runs contains a large amount of divalent elements e.g.  $\text{Mn}^{2+}$ ,  $\text{Ni}^{2+}$ , some  $\text{Fe}^{2+}$ , (as well as much  $\text{Fe}^{3+}$ ) and charge balance is achieved by entry of an equivalent amount of  $\text{Ti}^{4+}$  ( $\text{M}^{2+} + \text{Ti}^{4+} = 2\text{Al}^{3+}$ ). This phase is also capable of accepting a substantial amount of sodium although surprisingly, the complex substitution of ( $\text{Na}^{+} + \text{REE}^{3+}$ ) for  $2\text{Ca}^{2+}$  reported by Rockwell, was not observed. The hibonite produced in the present series of runs contained a large amount of  $\text{Fe}^{3+}$ . It is conceivable that hibonite could be a stable phase in SYNROC-D formulations prepared under more reducing conditions if there were sufficient  $\text{Ti}^{4+}$  to charge-compensate for  $\text{Fe}^{2+}$  ( $\text{Ti}^{4+} + \text{Fe}^{2+} = 2\text{Al}^{3+}$ ). This possibility should be investigated since hibonite could be a suitable host for sodium if the sodium content of the sludges could be reduced by more intensive washing. However, there are certain properties of hibonite which require additional study before it can be seriously considered as a component of SYNROC-D. These include the resistance to leaching of hibonites containing substantial amounts of sodium and the effects of radiation damage on this phase.

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Crystallization behaviour of interstitial glass  
in SYNROC-D formulations

1. INTRODUCTION

A new estimate of the composition of the average Savannah River sludge has recently become available. This report summarizes an experimental study of the incorporation of the new average sludge composition into SYNROC-D formulations and the crystallization behaviour of liquids in those systems. In some hot-pressing experiments carried out at 1100°C, a significant amount of interstitial glass was observed in addition to the major phases spinel ss., perovskite, zirconolite and nepheline. The presence of glass in SYNROC-D is undesirable, since it might serve as a host for cesium and strontium, and could lead to unacceptably high leach-rates for these elements. Accordingly, a further series of experiments was carried out in order to characterize the glass and to specify a process for its elimination. The recommended procedures are discussed in this report.

2. EXPERIMENTAL METHODS

Starting materials were prepared from oxide-hydroxide-carbonate mixtures following standard procedures as used in this laboratory. A total of 32 hot-pressing runs were carried out in a piston-and-cylinder apparatus at pressures of 5 kbar, and temperatures ranging from 1200°C to 900°C, over periods of 1-3 hours. The products were examined by X-ray diffraction, microscopic and microprobe techniques. Compositions of starting materials

were checked by microprobe bulk analysis of samples hot-pressed at subsolidus conditions ( $\sim 900^{\circ}\text{C}$ ).

### 3. INCORPORATION OF AVERAGE SAVANNAH RIVER SLUDGE INTO SYNROC-D

Table 1, column 1, presents the bulk composition of the new average sludge (after calcination), together with appropriate proportions of SYNROC additives ( $\text{CaO}$ ,  $\text{SiO}_2$ ,  $\text{TiO}_2$ ,  $\text{ZrO}_2$ ). This material was hot-pressed at  $1100^{\circ}$  and  $1200^{\circ}\text{C}$  in Ni capsules to produce the desired SYNROC mineralogy. The composition was further modified by the addition of more  $\text{SiO}_2$  such that the molar ratio of  $\text{SiO}_2:\text{Na}_2\text{O}$  was increased from an initial value of 2:1, to 2.5:1, 3:1 and 4:1. These three modified compositions were also hot-pressed in Ni at  $1100^{\circ}$  and  $1200^{\circ}\text{C}$ . The objective of these latter experiments was to study the effects of  $\text{SiO}_2$  activity upon the stability of SYNROC-D minerals, and on the composition of the glasses.

At  $1100^{\circ}$ , all four starting compositions (molar  $\text{SiO}_2:\text{Na}_2\text{O} = 2:1, 2.5:1, 3:1$  and  $4:1$  respectively) yielded assemblages consisting of magnetite-rich spinel, fine-grained perovskite and zirconolite, interstitial Al-rich glassy materials, and rare relicts of unreacted  $\text{ZrO}_2$  and  $\text{Al}_2\text{O}_3$ . Nepheline was not identified by microprobe, but was confirmed by X-ray diffraction.

In order to obtain a better approach to equilibrium a second series of hot-pressing runs was carried out at  $1200^{\circ}\text{C}$ . All compositions were found to contain large crystals of perovskite and aluminous spinel, and small crystals of nepheline. Zirconolite was too small for quantitative analysis, its presence being confirmed by X-ray diffraction. Analyses of coexisting phases are presented in Table 2. Textural features suggest the composition of the interstitial liquid has been modified by quenching, hence



Table 1. Bulk compositions of starting materials.

	"SC"	"High-neph SC"	"SYN-D Glass 1"	"SYN-D Glass 2"
SiO <sub>2</sub>	7.8	23.3	22.9	33.8
TiO <sub>2</sub>	18.8	15.3	11.3	8.6
ZrO <sub>2</sub>	5.6	5.0	0.3	1.4
UO <sub>2</sub>	2.2	1.9	-	-
Al <sub>2</sub> O <sub>3</sub>	19.8	22.8	29.0	25.4
Ce <sub>2</sub> O <sub>3</sub>	0.6	0.6 <sup>2</sup>	-	-
Nd <sub>2</sub> O <sub>3</sub>	0.6	-	-	-
Fe <sub>2</sub> O <sub>3</sub> <sup>1</sup>	25.3	9.5	17.2	9.8
MnO	5.1	2.0	3.4	2.0
NiO	3.3	0.9	1.7	-
CaO	7.3	7.1	5.3	4.6
SrO	0.3	0.3	- <sup>3</sup>	1.0
Na <sub>2</sub> O	3.5	12.0	8.7	1 <sup>~</sup>
Cs <sub>2</sub> O	0.3	1.0	1.0	1.0
Sum	100.0	101.0	100.0	101.0

<sup>1</sup> all Fe as Fe<sub>2</sub>O<sub>3</sub>.

<sup>2</sup> Sm<sub>2</sub>O<sub>3</sub> = 0.6%.

<sup>3</sup> 2% SrO subsequently added for partition experiments.

"SC" New average Savannah River s idge plus SYNROC additives.

"High-neph. S.C." Bulk composition of model SYNROC-D with 55 wt.% nepheline, 15 wt.% each perovskite, zirconolite and spinel.

"SYN-D Glass 1" Interstitial glass in "SC" composition at 1200°C.

"SYN-D Glass 2" Interstitial glass in "High-neph. S.C." at 1200°C.

Table 2. Coexisting phases in "SC", a SYNROC-D formulation designed to immobilize average Savannah River sludge.

	Spinel	Perovskite	Nepheline	Glass
SiO <sub>2</sub>	-	-	41.8	35.5
TiO <sub>2</sub>	7.1	53.3	-	9.4
ZrO <sub>2</sub>	-	<.1	-	0.3
UO <sub>2</sub>	-	2.1	-	<.1
Al <sub>2</sub> O <sub>3</sub>	29.9	-	37.1	23.7
Fe <sub>2</sub> O <sub>3</sub>	20.0	-	-	-
REE	-	~4	-	-
FeO	26.4	1.9	-	12.1
NiO	7.5	<.1	-	0.5
MnO	8.7	1.1	-	1.8
CaO	0.3	34.8	-	6.1
SrO	-	1.4	-	<.2
Na <sub>2</sub> O	-	1.2	21.0	9.8
Cs <sub>2</sub> O	-	-	<.2	<.2
Sum	99.8	99.8	99.9	99.2

Zirconolite too fine-grained for analysis.

Starting composition: as shown in Table 1.

Experimental conditions: 5 kbar, 1 hour, Ni capsule, 1200°C.

the analysed glass composition presented in Column 4 of Table 2, is at best an approximation of the liquid composition.

An important observation in these experiments is that nepheline has crystallized in bulk compositions whose bulk molar  $\text{SiO}_2\text{:Na}_2\text{O}$  ratios range from the 2:1 value normally prescribed for SYNROC-D through to very high values (4:1). This demonstrates that SYNROC-D wasteforms would be tolerant to fluctuations in the  $\text{SiO}_2\text{:Na}_2\text{O}$  ratio of the waste stream. In other words, large excess quantities of silica would not suppress nepheline crystallization. However, at a given temperature, excess quantities of silica cause an increase in the amount of interstitial glass.

It is significant that the glass compositions (Table 3) remain comparatively low in  $\text{SiO}_2$  and high in  $\text{Al}_2\text{O}_3$ , even in compositions with high  $\text{SiO}_2\text{:Na}_2\text{O}$  ratios. Apparently the liquid composition is buffered by equilibrium with crystalline phases, namely spinel ss, nepheline, perovskite and zirconolite. Glasses with these unusual compositions are likely to display higher leachabilities than, for example, basaltic aluminosilicate glasses with higher  $\text{SiO}_2$  contents. The presence of these glasses in SYNROC-D wasteforms will therefore be undesirable, because elements such as cesium and strontium which might partition into the glass could then be easily leached. On the other hand, the compositions of these glasses, namely low  $\text{SiO}_2$  and high  $\text{TiO}_2$ ,  $\text{FeO}$  and  $\text{Na}_2\text{O}$  suggest that they are likely to have low viscosities, and would crystallize readily with further cooling. Accordingly, a new series of experiments was designed to obtain a better estimate of the compositions of SYNROC-D glasses, so that their crystallization behaviour under controlled cooling conditions could be further evaluated.

Table 3. Glasses produced in "SC" compositions with variable molar  $\text{SiO}_2\text{:Na}_2\text{O}$  ratios<sup>1</sup>.

$\text{SiO}_2\text{:Na}_2\text{O}$ Temp.	3:1 1100 <sup>0</sup>	4:1 1100 <sup>0</sup>	2:1 1200 <sup>0</sup>	3:1 1200 <sup>0</sup>	4:1 1200 <sup>0</sup>
$\text{SiO}_2$	23.0	23.5	22.2	35.5	37.1
$\text{TiO}_2$	11.4	11.6	10.8	9.4	10.7
$\text{ZrO}_2$	0.4	0.2	0.3	0.3	<.1
$\text{Al}_2\text{O}_3$	29.4	26.8	23.9	23.7	22.3
$\text{FeO}^2$	17.8	14.8	13.8	12.1	11.1
$\text{NiO}$	1.6	2.6	0.9	0.5	0.4
$\text{MnO}$	3.7	4.1	2.5	1.8	2.4
$\text{CaO}$	4.3	7.2	4.3	6.1	7.8
$\text{Na}_2\text{O}$	7.8	8.1	10.1	9.8	7.5
$\text{Cs}_2\text{O, SrO}$	see footnote <sup>3</sup>				
Sum	99.4	98.9	98.8	99.2	99.3

<sup>1</sup> All runs in Ni capsules at 5 kbar.

<sup>2</sup> All iron expressed as  $\text{FeO}$

<sup>3</sup> Cesium volatilizes from the glasses under the impact of the electron beam during microprobe analysis. Accurate measurement of  $\text{Cs}_2\text{O}$  was not possible, however, the high sums (~99%) of all glass compositions suggest  $\text{Cs}_2\text{O}$  contents would be ~1%.

Analytical interferences arise in the determination of low levels of  $\text{SrO}$  in the presence of  $\text{SiO}_2$ . The  $\text{SrO}$  contents of the glasses are probably <0.5%.

#### 4. CHARACTERIZATION OF SYNROC-D INTERSTITIAL LIQUIDS

Accurate determination of the composition of the interstitial liquid in SYNROC-D is beset by many difficulties. Firstly, it is evident that the liquid composition is modified during quenching to glass by precipitation of crystallites. Secondly, the small percentage of glass lies along grain boundaries, in narrow bands, and does not segregate into pools large enough for satisfactory microprobe analyses. This second problem at least, can be overcome by a strategy familiar to experimental petrologists, whereby the proportion of glass relative to crystalline phases can be increased. It depends on the fact that the chemical equilibria between liquid and crystalline phases are independent of the proportions of each phase, providing that the compositions of the phases remain fixed. Accordingly, a modified form of SYNROC-D was prepared, in which the amount of interstitial glass should be greater than for the "SC" series of experiments. This modified composition, "High-neph SC" (Column 2, Table 1) corresponds to an assemblage of the typical SYNROC-D minerals, nepheline, zirconolite, perovskite and spinel ss, but in different proportions, namely 55 wt.% nepheline, and 15 wt.% each of the others.

It is evident from Table 3 that the interstitial SYNROC-D liquids contain large amounts of normative nepheline. It follows that in the "High-neph. SC" composition, in which the modal abundance of nepheline has been increased, the quantity of liquid in equilibrium with residual nepheline, spinel ss, zirconolite and perovskite will also be increased. In this manner, larger volumes of glass occur in the run products, and microprobe analysis is facilitated.

The "High-neph. SC" composition has a molar ratio of  $\text{SiO}_2$ :  
 $\text{Na}_2\text{O} = 2:1$ . A modification of this was prepared by adding excess

$\text{SiO}_2$  so that  $\text{SiO}_2:\text{Na}_2\text{O} = 3:1$ , in order to observe any change in glass composition as a result of higher  $\text{SiO}_2:\text{Na}_2\text{O}$  ratios.

These two nepheline-enriched compositions were hot-pressed in Ni capsules at  $1100^\circ$  and  $1200^\circ\text{C}$ . Reaction was incomplete at  $1100^\circ\text{C}$  because relict  $\text{ZrO}_2$  and  $\text{Al}_2\text{O}_3$  could be identified, together with perovskite, spinel and glass. In the 2:1 composition nepheline and zirconolite could also be identified. At  $1200^\circ\text{C}$  the product assemblages of both compositions (2:1 and 3:1) contained spinel, perovskite, zirconolite, nepheline and glass. Typical examples of the phase assemblages are presented in Tables 4 and 5. Although glass was much more abundant than in the "SC" series of experiments, some difficulties were still experienced in obtaining satisfactory microprobe analyses, because in many instances the glasses were peppered with tiny quench crystallites. It should be emphasised again that the glass analyses presented herein are at best an approximation of liquid compositions because of this problem. Table 6 contains analyses of glasses produced at  $1100^\circ$  and  $1200^\circ\text{C}$  in these nepheline-rich compositions.

As mentioned above, all glasses coexist with the major SYNROC-D phases (the high  $\text{Al}_2\text{O}_3$  content of the 1:2 glass at  $1100^\circ$  probably reflects incomplete reaction of  $\text{Al}_2\text{O}_3$ ). Certain chemical trends are evident as a function of increasing  $\text{SiO}_2$  content, and run temperature, namely higher  $\text{SiO}_2$ ,  $\text{TiO}_2$  and  $\text{CaO}$  and lower  $\text{Al}_2\text{O}_3$  contents. Although the starting material was doped with uranium and samarium, neither of these elements can be detected in the glass. This key observation suggests that even if an interstitial glass is produced during SYNROC-D manufacture, trivalent and tetravalent actinides and rare earths would nevertheless be contained in the crystal lattices of zirconolite and perovskite, and

Table 4. Coexisting phases plus glass in "High-neph. SC" with  $\text{SiO}_2\text{:Na}_2\text{O} = 2:1$  (Col. 2, Table 1).

	Perovskite	Zirconolite	Spinel	Nepheline	Glass
$\text{SiO}_2$	-	-	-	44.1	33.8
$\text{TiO}_2$	51.7	32.2	11.8	-	8.6
$\text{ZrO}_2$	0.2	40.3	-	-	1.4
$\text{UO}_2$	5.7	4.3	-	-	$\leq 0.2$
$\text{Sm}_2\text{O}_3$	$\sim 4.0$	$\sim 1.0$	-	-	-
$\text{Al}_2\text{O}_3$	1.6	2.2	23.5	34.7	25.4
$\text{Fe}_2\text{O}_3$	-	-	18.9	-	-
$\text{FeO}$	1.4	6.6	30.1	-	8.8
$\text{NiO}$	$< 1$	0.3	8.9	-	0.2
$\text{MnO}$	0.7	1.8	7.1	-	2.0
$\text{CaO}$	32.2	10.8	0.3	-	4.6
$\text{Na}_2\text{O}$	1.3	$\sim 0.3$	-	18.8	13.7
$\text{Cs}_2\text{O}$	-	-	-	1.4	0.2
Sum	100.3*	99.6	100.6	99.0	98.7

\*includes 1.5% SrO.

Starting material: "High-neph. SC" (Column 2, Table 1).

Experimental conditions: 5 kbar,  $1200^\circ\text{C}$  1 hr, Ni capsule.

Note: A fine-grained Cs-rich phase may be present, but positive microprobe identification could not be achieved. A pollucite phase ( $\text{CsAlSi}_2\text{O}_6$ ) has not been positively identified in any of these experiments. It is possible that pollucite is not in equilibrium with the highly  $\text{SiO}_2$ -undersaturated liquids produced in these experiments.

Table 5. Coexisting phases plus glass in "High-neph. SC", with  $\text{SiO}_2:\text{Na}_2\text{O} = 3:1$ .

	Perovskite	Zirconolite	Spinel	Glass*
$\text{SiO}_2$	-	-	-	41.7
$\text{TiO}_2$	52.9	33.6	19.6	10.3
$\text{ZrO}_2$	0.2	41.1	-	1.2
$\text{UO}_2$	4.3	2.4	-	~0.2
$\text{Sm}_2\text{O}_3$	~2.	~1.	-	-
$\text{Al}_2\text{O}_3$	1.3	1.8	4.6	22.5
$\text{Fe}_2\text{O}_3$	-	-	25.3	-
$\text{FeO}$	1.6	8.1	27.4	7.9
$\text{NiO}$	<.1	0.2	15.9	0.2
$\text{MnO}$	0.4	0.3	6.8	1.8
$\text{CaO}$	35.6	10.6	0.3	6.7
$\text{Na}_2\text{O}$	1.6	~0.4	-	6.3
$\text{Cs}_2\text{O}$	-	-	-	1.0
Sum	100.1	99.5	100.0	99.6

Nepheline occurs as numerous small inclusions in perovskite and zirconolite, and also in fine-grained intergrowths with spinel in the interstitial glass.

Starting material: "High-neph. SC" as given in Column 2, Table 1 plus additional  $\text{SiO}_2$  so that  $\text{Na}_2\text{O}:\text{SiO}_2 = 1:3$ .

Experimental conditions: 5 kbar,  $1200^\circ$ , 1 hour, Ni capsule.

\*Analysis may not be very meaningful because of beam overlap onto nepheline and spinel.



Table 6. Analyses of glasses produced<sup>1</sup> in "High-neph. SC" compositions with molar  $\text{SiO}_2:\text{Na}_2\text{O} = 1:2$  and  $1:3$ .

$\text{SiO}_2:\text{Na}_2\text{O}$ Temp. C°	1:2 1100°	1:3 1100°	1:2 1200°	1:3 1200°
$\text{SiO}_2$	23.3	34.4	33.8	41.7
$\text{TiO}_2$	1.9	9.4	8.6	10.3
$\text{ZrO}_2$	0.3	2.4	1.4	1.2
$\text{Al}_2\text{O}_3$	47.8	26.2	25.4	22.5
$\text{FeO}^{2,3}$	8.2	7.1	8.8	7.9
$\text{NiO}$	2.2	0.9	0.2	0.2
$\text{MnO}$	2.5	1.7	2.0	1.8
$\text{CaO}$	1.4	4.8	4.6	6.7
$\text{Na}_2\text{O}$	11.4	11.9	13.7	6.3
$\text{Cs}_2\text{O}, \text{SrO}$	see footnote <sup>3</sup> .....			
Sum	99.3	98.8	98.7	99.6

<sup>1</sup> All runs in Ni capsules, 5 kbar, 1 hour.

<sup>2</sup> All iron as  $\text{FeO}$ .

<sup>3</sup> Some  $\text{Cs}_2\text{O}$  (~1%) is present in the glass, but this volatilizes under the electron beam, making quantitative analysis impossible.  $\text{SrO}$  is probably  $\leq 0.5\%$ .

would not partition into the glass. Leach-rates for these elements should therefore remain low.

There are analytical difficulties in determining small abundances of SrO in the presence of SiO<sub>2</sub>, nevertheless, there appears to be about 0.5% SrO in the glass. In glasses, cesium volatilizes under the electron beam of the probe, making analysis difficult, however, it seems that most of these glasses contain about 1% or more of Cs<sub>2</sub>O.

##### 5. CONTROLLED COOLING AND CRYSTALLIZATION OF SYNROC-D GLASSES

Estimates of the composition of interstitial liquids in SYNROC-D have been obtained as described in Sections 3 and 4 and are summarised in Tables 3 and 6. Two such compositions were selected for controlled cooling and crystallization experiments (Cols. 3 and 4, Table 1). These compositions, "SYN-D Glass 1" and "SYN-D Glass 2" correspond to the glasses produced during hot-pressing at 1200°C in, respectively, "SC" composition, and "High-neph. SC, SiO<sub>2</sub>:Na<sub>2</sub>O = 2:1". The suitability of these two compositions as model interstitial liquids was verified by X-ray diffraction study of both compositions following devitrification at 900°C. In both cases, the SYNROC-D phases spinel ss, perovskite and nepheline were observed. (The low bulk abundance of ZrO<sub>2</sub> in these compositions precludes identification of zirconolite). Furthermore, in subsequent melting experiments these same phases were found to crystallize from both compositions. This evidence of saturation with SYNROC-D phases likewise confirms the suitability of using "SYN-D Glass 1" and "SYN-D Glass 2" compositions as model interstitial liquids.

The "SYN-D Glass 1" composition was cooled linearly from

1100° to 900°C over 2 hours in Fe capsules, and similar cooling runs were conducted from 1200° to 900°C in both Fe and Pt capsules. The product assemblages consisted of nepheline, spinel ss, perovskite, and possible residual glass in the Pt-encapsulated run. Analytical data in Table 7 show that cesium is taken up by nepheline as it precipitates (mass-balance considerations indicate that trace quantities of another Cs-rich phase may also have been precipitated).

The partition behaviour of strontium was studied in a separate experiment. Analytical interferences pose problems for the detection of small amounts of strontium in silicate glass; but these were overcome by doping the starting material (Col. 1, Table 1) with ~2% SrO and cooling it in Fe capsules from 1200° to 1050°C in order to obtain a product assemblage of nepheline, spinel, perovskite and glass. The perovskite was found to contain 4-5% SrO whilst the glass contained less than 0.5% SrO. The pronounced partitioning of Sr into perovskite is most encouraging. It suggests that most or all of the strontium in interstitial liquids which develop during hot-pressing of SYNROC-D at 1000-1200°C could be incorporated into the relatively leach-resistant perovskite phase by slow-cooling to 800-900°C. Likewise, most of the cesium in the glass could be incorporated into crystalline phases, mainly nepheline.

The other model liquid composition, "SYN-D Glass 2" (Col. 4, Table 1) was similarly cooled from 1200° to 900°C in Fe and Pt capsules. In Fe capsules perovskite, spinel, rare zirconolite and two types of nepheline were identified (Table 8). X-ray diffraction confirmed that nepheline was very abundant. The two types of nepheline differ in that type (b) contains more Si, Ti, Fe, Mn, and Cs than type (a) and probably represents a late-

Table 7. Coexisting phases precipitated as a result of slow-cooling of "SYN-D Glass 1".

(A) Assemblage produced in Fe capsule

	Spinel	Perovskite	Nepheline
$\text{SiO}_2$	-	-	40.1
$\text{TiO}_2$	26.2	57.2	0.8
$\text{Al}_2\text{O}_3$	12.3	0.4	36.6
$\text{Fe}_2\text{O}_3$	4.2	-	-
FeO	49.4	1.3	1.0
NiO	<.1	<.1	-
MnO	8.1	0.9	0.2
CaO	0.1	39.8	0.7
$\text{Na}_2\text{O}$	-	0.5	17.3
$\text{Cs}_2\text{O}$	-	-	2.6
Sum	100.3	100.1	99.3

(B) Assemblage produced in Pt capsule

	Spinel	Perovskite	Nepheline
$\text{SiO}_2$	-	-	41.3
$\text{TiO}_2$	10.6	56.1	0.7
$\text{Al}_2\text{O}_3$	5.1	1.5	36.6
$\text{Fe}_2\text{O}_3$	42.3	-	-
FeO	27.9	1.4	0.6
NiO	4.9	<.1	-
MnO	8.8	0.9	0.3
CaO	0.2	38.8	0.5
$\text{Na}_2\text{O}$	-	1.3	19.3
$\text{Cs}_2\text{O}$	-	-	1.1
Sum	99.8	100.0	100.4

Starting composition: "SYN-D Glass" Col. 3, Table 1.

Experimental conditions: 5 kbar,  $1200^\circ$  for 15 mins, then cool to  $900^\circ\text{C}$  over 3 hours.

Table 8. Phase assemblages in slow-cooled "SYN-D Glass 2".

(A) Assemblage produced in Fe capsule

	Pvsk	Sp.	Zirc.	Nepheline	
				(a)	(b)
SiO <sub>2</sub>	-	-	-	42.2	43.2
TiO <sub>2</sub>	56.8	31.3	34.5	0.1	0.7
ZrO <sub>2</sub>	0.2	0.2	45.6	-	-
Al <sub>2</sub> O <sub>3</sub>	0.2	4.8	1.7	36.4	32.6
FeO	1.3	56.0	4.5	0.3	2.3
MnO	0.7	7.4	1.5	-	0.6
CaO	36.4	0.1	10.2	1.2	2.5
SrO	4.2	-	1.2	-	-
Na <sub>2</sub> O	0.6	-	0.4	19.3	16.7
Cs <sub>2</sub> O	-	-	-	0.4	1.1
Sum	100.4	99.8	99.6	99.9	99.7

(B) Assemblage produced in Pt capsule

	Pvsk	Sp.	Zirc.	Neph.
SiO <sub>2</sub>	-	-	-	43.7
TiO <sub>2</sub>	57.1	23.2	30.7	0.6
ZrO <sub>2</sub>	<.1	<.1	44.2	<.1
Al <sub>2</sub> O <sub>3</sub>	0.5	1.6	2.4	34.5
Fe <sub>2</sub> O <sub>3</sub>	-	21.9	-	-
FeO	1.1	43.8	9.1	2.0
MnO	0.9	8.6	2.2	0.4
CaO	36.7	0.2	10.2	2.1
SrO	3.6	-	0.3	-
Na <sub>2</sub> O	0.6	-	0.5	16.3
Cs <sub>2</sub> O	-	-	-	0.4
Sum	100.5	99.3	99.6	100.0

Experimental conditions: 5 kbar, 1200° for 15 mins, then cool to 900°C over 3 hours.

crystallizing or quench-produced nepheline. Accordingly, the "cleaner" type (a) nepheline is probably an early-crystallizing phase. Cooling experiments in Pt capsules produced an assemblage of perovskite, spinel, rare zirconolite and type (b) nepheline. No glass was identified in either product assemblage. Mass balance considerations for cesium suggest that trace amounts of a Cs-rich phase may also be present.

#### 6. LEACHING BEHAVIOUR OF HEAT-TREATED SYNROC-D

In view of the preceding evidence concerning the readily-crystallized nature of glasses formed during partial melting of SYNROC D, it seemed desirable to investigate the effect of different cooling rates and annealing temperatures on the leachability of SYNROC-D.

The sample chosen for investigation was S-21, which had previously been prepared by LLL and sent to Savannah River for evaluation. Petrographic studies showed that this sample contained interstitial glass along grain boundaries. Wet-way analyses gave 3.39%  $\text{Na}_2\text{O}$  and 9.48%  $\text{SiO}_2$ , i.e. a molar  $\text{SiO}_2/\text{Na}_2\text{O}$  ratio of 3.5. Formation of the glass had evidently been encouraged by this large ratio and possibly by a hot-pressing temperature somewhat higher than intended. The substantial Sr, Ca and Al leach rates seen in S-21 were probably due to selective leaching of these elements from the glass.

Chips of S-21 were wrapped in platinum foil and placed in evacuated silica tubes. Sample A was heated at  $900^\circ\text{C}$  for 3 hours; Sample B was heated at  $1075^\circ\text{C}$  for 30 minutes and then cooled to  $900^\circ\text{C}$  over 3 hours; Sample C was heated at  $1160^\circ\text{C}$  for 30 minutes and then cooled to  $900^\circ\text{C}$  over 3 hours. After heat treatment,

the samples were leached in deionized water at 95°C and leachates analyzed for Sr and Ca following established procedures at this laboratory. Results are given in Table 9.

It is seen that the heat treatments have made little difference to the leach rates. This result is surprising since experiments show that any interstitial glass should have become fully crystallized by the heat-treatments to which they were subjected, and that Sr and Ca should have partitioned quantitatively into the perovskite phase, thereby causing a decrease in leach rates.

Further experiments will be necessary to clarify the situation. One possibility is that the effective surface areas of devitrified interstitial glass are increased by crystallization under vacuum and that this effect cancels out the expected decrease in leach rate caused by crystallization of perovskite. This interpretation will be tested by performing experiments on samples which are cooled under pressure. The effects of differing cooling rates should also be explored. Investigations should also be made of the structure of devitrified glass regions by electron microscopy.

## 7. CONCLUSION

The interstitial glass formed by partial melting of SYNROC-D at 1100-1200°C is silica-poor and alumina-rich. These characteristics are maintained when large variations in  $\text{SiO}_2/\text{Na}_2\text{O}$  ratios are imposed, provided that the liquids are buffered by the main SYNROC crystalline phases- spinel, nepheline, perovskite and zirconolite. However, the proportion of glass increases as the  $\text{SiO}_2/\text{Na}_2\text{O}$  ratio increases. Glasses of this type possess quite low viscosities and experimental studies show that they are readily

Table 9. Leach rates in S-21 SYNROC-D subjected to various heat-treatments. Units are in multiples of  $10^{-5}$  g/cm<sup>2</sup>.day.

Day	<u>Strontium</u>			
	A	B	C	D (pristine sample)
1-3	3.0	2.0	1.9	1.0
4-6	3.7	2.0	2.3	1.9
7-12	1.9	1.1	1.1	1.4
12-17	1.9	1.9	2.0	
18-22	4.5	2.2	1.9	
	<u>Calcium</u>			
	A	B	C	D
1-3	1.3	1.2	1.0	1.4
4-6	1.4	0.8	1.0	1.1
7-12	0.8	0.4	0.5	1.1
12-17	0.9	0.8	0.9	
18-22	2.0	1.0	0.9	



crystallized by controlled cooling. The principal phases crystallizing are nepheline, perovskite and spinel, as would be expected for liquids in equilibrium with the SYNROC-D phase assemblage. During crystallization, strontium and calcium are strongly partitioned into the perovskite phase.

The unexpectedly high leachabilities of Ca, Sr and Al seen in S-21 were probably due to selective leaching of these elements from the glass. Surprisingly, the leach rates were not substantially decreased by heat-treating the samples in vacuo so as to devitrify the glass phase. This may have been due to an effective increase of surface area during devitrification. It is hoped that further studies of devitrification at different cooling rates under confining pressure will lead to improved leaching characteristics for calcium and strontium.

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