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Removal of Uranium from Spent Salt from the Molten Salt Oxidation Process

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Introduction

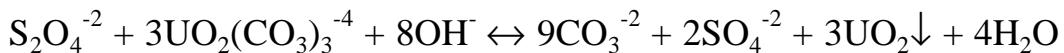
Molten salt oxidation (MSO) is a thermal process that has the capability of destroying organic constituents of mixed wastes, hazardous wastes, and energetic materials. In this process, combustible waste and air are introduced into the molten sodium carbonate salt. The organic constituents of the waste materials are oxidized to carbon dioxide and water, while most of the inorganic constituents, including toxic metals, minerals, and radioisotopes, are retained in the molten salt bath. As these impurities accumulate in the salt, the process efficiency drops and the salt must be replaced. An efficient process is needed to separate these toxic metals, minerals, and radioisotopes from the spent carbonate to avoid generating a large volume of secondary waste. Toxic metals such as cadmium, chromium, lead, and zinc etc. are removed by a method described elsewhere. This paper describes a separation strategy developed for radioisotope removal from the mixed spent salt, as well as experimental results, as part of the spent salt cleanup.

As the MSO system operates, inorganic products resulting from the reaction of halides, sulfides, phosphates, metals and radionuclides with carbonate accumulate in the salt bath. These must be removed to prevent complete conversion of the sodium carbonate, which would result in eventual losses of destruction efficiency and acid scrubbing capability. There are two operational modes for salt removal: (1) during reactor operation a slip-stream of molten salt is continuously withdrawn with continuous replacement by carbonate, or (2) the spent salt melt is discharged completely and the reactor then refilled with carbonate in batch mode. Because many of the metals and/or radionuclides captured in the

salt are hazardous and/or radioactive, spent salt removed from the reactor would create a large secondary waste stream without further treatment. A spent salt clean up/recovery system is necessary to segregate these materials and minimize the amount of secondary waste. These materials can then be encapsulated for final disposal.

Uranium Removal by Chemical Reduction and Precipitation

When the concentration of carbonate in the spent salt is low (as in mode 2) most of the mineral residues or metal species have limited solubility (if the pH is still >8) and will precipitate as metal oxides or hydroxides. The precipitation can be enhanced by NaSH which forms sulfides of the mineral species, thereby further reducing the solubility¹. The addition of other reagents may be required to convert the valence state of dissolved metals into insoluble forms. Dithionite ($\text{Na}_2\text{S}_2\text{O}_4$) reduces soluble Cr^{+6} to Cr^{+3} , forming insoluble $\text{Cr}(\text{OH})_3/\text{Cr}_2\text{O}_3$ which has low solubility in alkaline solutions. Dithionite will also reduce uranium (VI) to uranium (IV), which is insoluble, and precipitates. The precipitated U(IV) can be removed by filtration. The following equation describes this reaction:



The results from experiments using dithionite ($\text{Na}_2\text{S}_2\text{O}_4$) to precipitate uranium from low carbonate solution, at room temperature, are shown below in Table 1: (Uranium concentration was measured using ICP-MS).

Table 1:

<u>Exp. #:</u>	<u>Initial U conc.</u>	<u>U conc. after Dithionite treatment</u>
U Red 1	164 ppm	0.235 ppm
U Red 2	2220 ppm	0.790 ppm

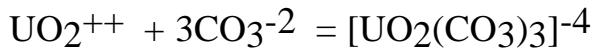
In the presence of 50 mole% (65 wt%) Na_2CO_3 the dithionite is ineffective in reducing U(VI) to U(IV), as the experimental results in Table 2 show:

Table 2:

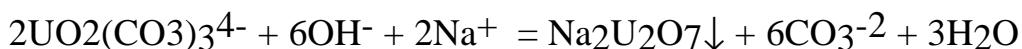
<u>Exp. #:</u>	<u>Initial U conc.</u>	<u>U conc. after Dithionite treatment</u>
U Red 3	2568 ppm	2313 ppm

Radionuclides such as uranium form a highly soluble complex in carbonate solutions. Treatment with alkali (NaOH) forms an insoluble sodium diuranate ($\text{Na}_2\text{U}_2\text{O}_7$)² which is recovered by filtration, yielding ppm levels of uranium in solution. The carbonate solution is then dried and recycled back to the MSO reactor for reuse. If the spent salt contains low levels of carbonate and is destined for disposal, the uranium can be removed by a chemical reduction method (such as dithionite or NaSH), similar to that used for metals precipitation³. Any residual uranium in solution is further removed using an ion exchange resin, Diphonix®, which has a high selectivity for uranium over a specific pH range. This paper summarizes the experiments demonstrating the alkali treatment technique to reduce uranium concentrations to 50 ppm levels, and the use of an ion exchange resin to reduce the uranium further to sub ppm levels.

The dissolved spent salt from the molten salt reactor, in the presence of carbonate and oxygen, contains uranium in the hexavalent (uranyl) state as sodium uranyl tricarbonate.



Under these conditions, at a pH of about 10-12, many elements are precipitated as hydroxides or carbonates (Fe, Mn, Al, Pb, Ca) while uranium remains in solution as the highly soluble sodium uranyl tricarbonate complex⁴. Using techniques derived from early uranium refining techniques we discovered that uranium is precipitated by the addition of sodium hydroxide. When the hydroxyl ion concentration becomes great enough, (pH>11) or an excess of sodium hydroxide of about five grams per liter, sodium diuranate will precipitate:⁵



The following test results in Table 3 illustrate the removal of uranium by the addition of NaOH to a 50 mole percent Na₂CO₃, NaCl solution.

Table 3:

<u>Initial U conc.</u>	<u>U conc. after NaOH to pH 12.5</u>	<u>U conc. after NaOH to pH 13.2</u>
2453 ppm	190.51 ppm	46.7 ppm

Further uranium removal is accomplished using an ion exchange resin, specific for uranium.

Ion Exchange for Uranium Removal

Several ion exchange resins were evaluated for efficiency of uranium removal; Diphonix® (developed by Eichrom Industries, Inc.)⁶, and NaTiO (from Boulder Scientific, Inc.). Initial efficiency tests were performed by placing 0.1 g of test resin in a 20 cc syringe, (loading capacity of 20%). A kynar frit (30 um porosity) was placed at the bottom of each syringe. Each syringe, containing 20 ml. of test solution, was placed on a rotator/mixer at 1 rev/4 sec. The pH of each test solution was adjusted, using HCl, to 5.5 (the recommended optimum) prior to contact with each resin. The results are displayed in Table 4.

Table 4:

Experiments at pH 5.5:

(exp. #'s IE-4,5)

<u>% Column Loading*</u>	<u>Initial U conc.</u>	<u>U conc. after Diphonix®</u>	<u>U conc. after NaTiO</u>
20	0.8 ppm	0.10 ppm	--
20	1.2 ppm	0.12 ppm	6.5 ppm
40	4.7 ppm	0.38 ppm	30.1 ppm
80	76.9 ppm	0.52 ppm	10.3 ppm

* Diphonix loading capacity is 40 mg/g resin

The above results (Table 4) reflect the resin performance at the recommended optimum pH (5.5). We also evaluated each resin using the same protocol as described above, at pH 10.0, to measure the resin performance at a pH closer to that of the filtrate following NaOH treatment.

Table 5:

Experiments at pH 10.0:

<u>Exp. #:</u>	<u>Initial U conc.</u>	<u>U conc. after Diphonix®</u>	<u>U conc. after NaTiO</u>
IE-8	51.1 ppm	29.8 ppm	38.6 ppm
IE-4	86.7 ppm	69.1 ppm	80.5 ppm
IE-4	179.8 ppm	150.0 ppm	160.4 ppm

Diphonix® resin has much better affinity for uranium than does NaTiO, and as expected the affinity is better at the optimum pH of 5.5. The Diphonix® resin was then tested (using the same protocol as above) at pH 7.0 and 8.5 to evaluate the affinity at pH's just above optimum. The results are shown in Table 6.

Table 6: (data from Exp IE-7)

<u>pH</u>	<u>Initial U conc.</u>	<u>U conc. after Diphonix®</u>
7.0	41.9 ppm	3.7 ppm
7.0	85.0 ppm	3.9 ppm
7.0	165.0 ppm	8.8 ppm
8.5	41.6 ppm	13.5 ppm
8.5	85.0 ppm	15.8 ppm
8.5	170.0 ppm	58.3 ppm

As the above results indicate, the affinity of Diphonix® resin for uranium, above the optimum pH of 5.5, is still relatively good.

Table 7 and Figure 1 summarize the effect of solution pH on the uranium removal efficiency with Diphonix® resin. Optimal performance of the resin occurs at pH 5.5.

Table 7:

Initial U conc. (ppm)	U conc. after Diphonix® at various pH			
	5.5	7.0	8.5	10.0
46.1 (41.2~51.1)	0.12	3.7	13.5	29.8
85.7 (84.7~86.7)	0.38	3.9	15.8	69.1
170.0 (160.0~170.0)	0.52	8.8	58.3	150.0

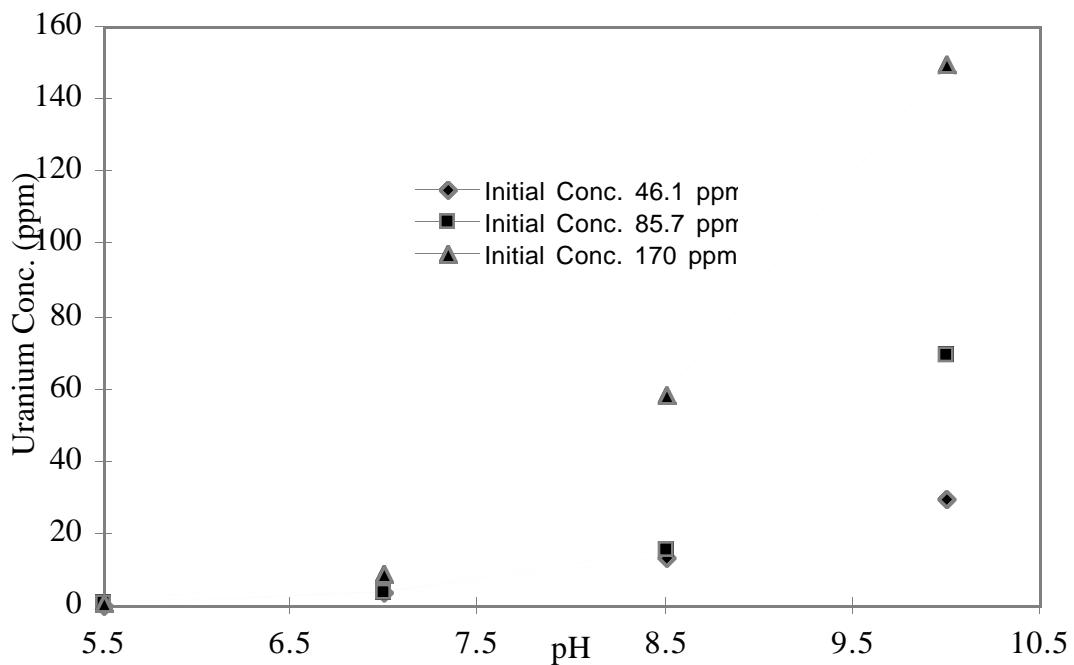


Fig. 1. Effect of Solution pH on Uranium Removal Efficiency with Diphonix® Resin

Diphonix Resin® for Uranium Removal in a Circulating System

Using the information obtained in the preceding experiments we then designed several experiments to test the affinity of Diphonix® in a set up similar to that designed for the MSO pilot plant. The Diphonix® is placed in an ion exchange column and the test solution is pumped around at a particular rate. Periodically, grab samples are removed from the feed tank to measure concentration (by ICP-MS). Over time, the uranium concentration in the feed tank should decrease, as uranium is removed by the column. The schematic for this set up is shown below:

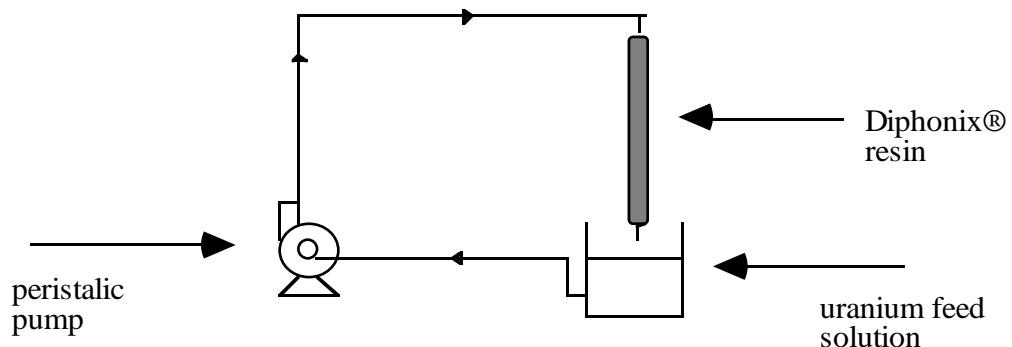


Fig. 2 Uranium removal using Diphonix® resin

Below (in Table 8) are the results of experiments modeling the pilot plant design, in which a carbonate free uranium feed solution (at a starting concentration of ~ 150 ppm) is pumped through the ion exchange column containing Diphonix® at a flow rate of 30 cc/min and 10 cc/min, loading of 16%, and a pH of 5.5.

Table 8

Time	Uranium conc. (ppm)	
	@ 30 cc/min	@ 10 cc/min
0 hr	146	144
1 hr	58	61
2 hr	43	39
3 hr	35	30
4 hr	29	26
5 hr	26	24
6 hr		23
8 hr	20	22
10 hr		23
12 hr	17	

The results, depicted graphically in Figure 3, indicate that a carbonate free feed solution of uranium, at a concentration of ~150 ppm, could be reduced by ion exchange to 10-20 ppm

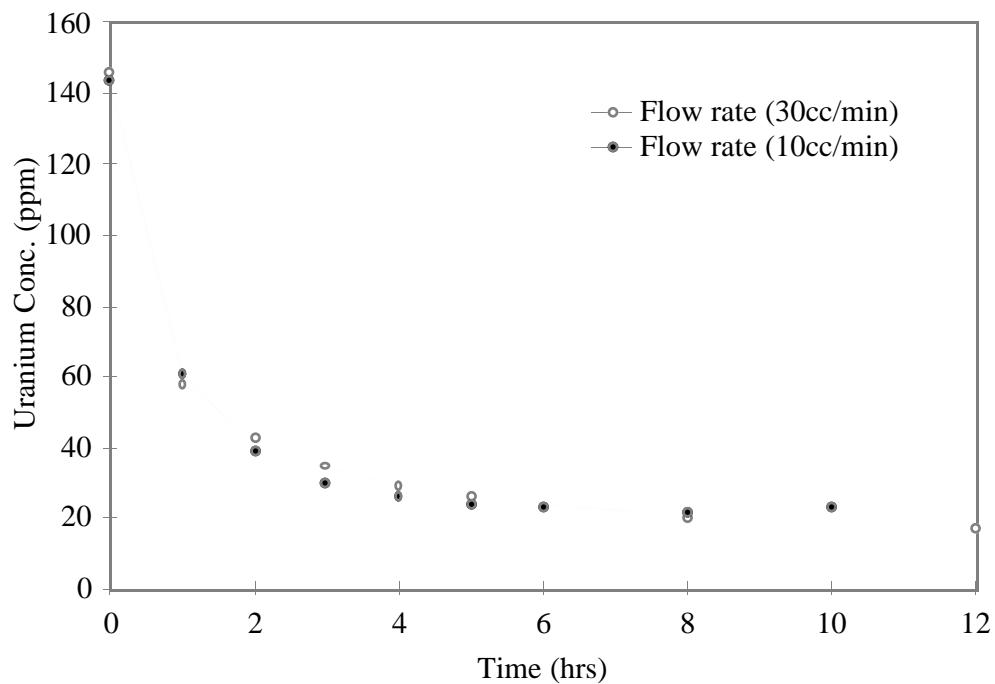


Fig. 3 Effect of Flow rate on Uranium Retension with Diphenix® Resin

The more realistic scenario would be a waste stream which was chemically pretreated with either dithionite ($\text{Na}_2\text{S}_2\text{O}_4$) or NaOH to remove most of the uranium by chemical precipitation and then polishing with ion exchange.

The following results show the excellent performance of Diphenix® as a polishing step to remove residual uranium. The feed solution, at a starting uranium concentration of 7 ppm, was pumped through the resin at 10 cc/min, with a loading capacity of 16%, and the pH of the feed solution adjusted to 5.5.

Table 9:

<u>Time</u>	<u>Uranium conc. (ppm)</u>
0 hr	7.05
0.3 hr	2.10
0.6 hr	0.97
1.0 hr	0.27
1.5 hr	0.032
4.0 hr	< 0.010
6.0 hr	< 0.010

We also evaluated the loading capacity of the Diphenix® ion exchange resin to determine at what concentration uranium would breakthrough the column and exceed the loading capacity. These were followed by elution studies using $(\text{NH}_4)_2\text{CO}_3$ to remove the uranium from the spent column. The results are displayed in Table 10. (Loading capacity 80%, flow rate 30 cc/min, pH of feed solution 5.5, uranium starting concentration 734 ppm.)

Table 10:

Time	Uranium conc. (ppm)
0 hr	734
0.3 hr	543
0.6 hr	486
1.0 hr	439
4.0 hr	345
8.0 hr	349
14.5 hr	387

Breakthrough appears at about 350 ppm Uranium under these conditions. We eluted the column in the previous experiment, using 8 bed volumes of 2M $(\text{NH}_4)_2\text{CO}_3$ and were able to recover about 89% of the uranium retained on the column. This will permit reuse of the column resin and a substantial cost saving.

Summary:

The preceding experiments demonstrate a useful separation strategy using chemical reduction with $\text{Na}_2\text{S}_2\text{O}_4$ or precipitation with NaOH and use Diphonix® ion exchange resin to remove uranium from the Molten Salt waste stream to sub ppm levels. This method also has application to the clean up of many environmental aqueous wastes. Further studies will evaluate the performance of these methods on Thorium waste streams.

Acknowledgments

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References:

- 1 "EQ 3/6, A Software Package for Geochemical Modeling of Aqueous Systems", Thomas J. Wolery, UCRL-MA-11062, Sept. 1992.
- 2 "The carbonate Chemistry of Uranium: Theory and Applications", McClaine, L.A., Bullwinkel, E.P., and Huggins, J.C., Proceedings of the International Conf. on Peaceful Uses of Atomic Energy, Aug 8-20, (1955), USAEC Report, Vol. VIII, P/525.
3. "Spent salt Clean-up/Recovery Process Development for Molten Salt Oxidation", Hsu, Peter C., von Holtz, Erica, Hipple, David, Summers, Leslie, and Brummond, William. in press.
- 4 Analytical Chemistry of Uranium, Palei, P.N. p 251, 1970.
- 5."Alkaline Leaching of Uranium Ores", Stephens, F.M.Jr., Macdonald, R.D., Proceedings of the International Conf. on Peaceful Uses of Atomic Energy, Aug 8-20, (1955), USAEC Report, Vol.VIII, P/520.
6. "Diphonix® Resin: A Review of its Properties and Applications", Chiarizia, R., Horwitz, E.P., Alexandratos, S.D., and Gula, M.J.. Separation Science and Technology (in press) (1996).

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