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REFINING OF URANIUM FROM LEACH PLANT ION EXCHANGE NITRATE ELUATES BY EXTRACTION WITH DIBUTYL BUTYLPHOSPHONATE

by

V. M. MCNAMARA, H. W. PARSONS & R. SIMARD

EXTRACTION METALLURGY DIVISION

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Mines Branch Investigation Report IR 60-120

REFINING OF URANIUM FROM LEACH PLANT ION EXCHANGE NITRATE ELUATES BY EXTRACTION WITH DIBUTYL BUTYLPHOSPHONATE

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V.M. McNamara*, H.W. Parsons*, and R. Simard**

SUMMARY OF RESULTS

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Continuous liquid-liquid extraction tests were done on current samples of ion exchange nitrate eluates with 25% v/vdibutyl butylphosphonate in kerosene. From eluates containing 15 to 28 g $U_3O_8/1$, recoveries of 99% or better were obtained in four countercurrent stages. The effects of nitrate and sulphate concentrations on extraction equilibria were determined. The extract, analyzing 75 to 95 g $U_{3}O_{8}/1$, was stripped effectively in four stages with 5 to 10% ammonium sulphate solutions at pH's ranging from 3.5 in the first stage. to 6.0 in the last. The uranium was precipitated as ammonium diuranate by gaseous ammonia, and the barren strip was recycled to the circuit. The effect of temperature and dilution of the ammonia with nitrogen on the filtering rate of the ammonium diuranate (ADU) precipitate and the subsequent reduction and sintering properties were investigated. The product was of nuclear grade purity and was used to produce high density (>10.5) uranium dioxide pellets.

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INTRODUCTION

In the established flowsheet of most uranium acid leach plants, uranium is recovered from leach solutions by anion exchange resins which normally are eluted with acidified nitrate solutions. The eluate is then neutralized in two stages, the first to pH 3.5 with lime, and the second to pH 7.0 with ammonia, magnesia or caustic soda. The first-stage precipitate or iron cake is recycled to leaching while the final uranium yellow cake is shipped to a custom refinery for reextraction with tributyl phosphate (TBP) and production of nuclear grade uranium trioxide or ammonium diuranate.

Previous studies⁽¹⁾⁽²⁾ have been made at the Mines Branch on the possibilities of refining the uranium by direct extraction with TBP from nitrate eluates, or by the sulphuric acid elution-amine extraction process. A third possibility was suggested by the availability of data⁽³⁾ on the properties of dialkyl alkylphosphonates as efficient extractants of uranyl nitrate from nitric acid solutions. It was considered likely that phosphonates would be more suitable than TBP, because of the higher extraction coefficients obtained at the lower nitric acid concentration used for ion exchange elution (0.3-0.4 N HNO₃).

Two kerosene-soluble phosphonates are available commercially in Canada;* they are dibutyl butylphosphonate (DBBP) and bis-2 ethylhexyl 2-ethylhexylphosphonate. These were first investigated in batch-scale

* Electric Reduction Co. of Canada, Toronto, Ontario.

tests in the laboratory, and, on the basis of the better performance of the former, it was then decided to carry out continuous bench-scale tests with nitrate eluates obtained from operating plants.

Batch equilibrium tests are continuing and will be reported separately. These tests had already defined certain operating conditions, mainly the uranium capacity of 25% DBBP and the necessity of using ammonium sulphate for effective stripping, but the continuous tests were necessary 1) to confirm extraction and stripping equilibria, 2) to determine the selectivity of the extractant as evidenced by the purity of the final product, 3) to determine the solubility and entrainment losses of phosphonate, and 4) to demonstrate the adaptability of such a process to current leach plant circuits.

For this work, drum quantities of eluate were obtained from Stanleigh Uranium Mines Limited and Consolidated Denison Mines Limited, of Elliot Lake, Ontario. A lesser quantity of nitrate eluate was obtained from operation of the ion exchange pilot plant (4 in. diameter columns) assembled at Ottawa by the Extraction Metallurgy Division of the Mines Branch. The feed to the ion exchange pilot plant was a solution produced by acid leaching of ore obtained from Milliken Lake Uranium Mines Ltd., Elliot Lake, Ontario.

PROCEDURE

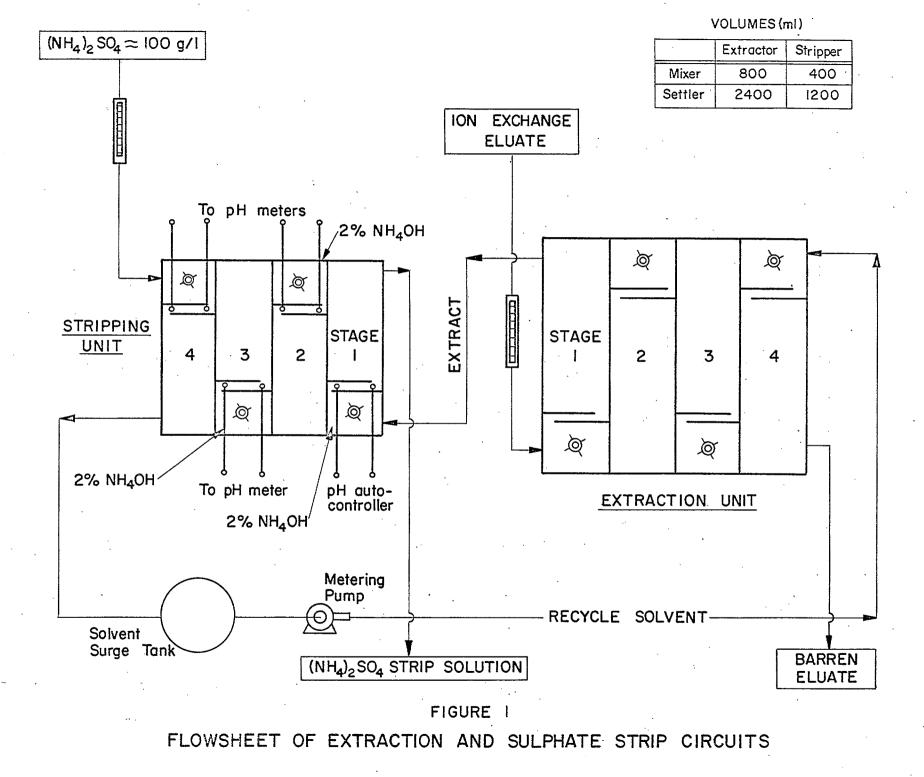
Extraction and Stripping

The existing four-stage mixer-settler units used in earlier tests were used for this work. The reasons for the choice of this equipment over pulse columns are discussed in a previous report.(1)

The flowsheet shown in Figure 1 was adopted for all tests. Each lot of eluate was first neutralized with lime to pH 3.0 to 3.5, filtered and reacidified with nitric acid to 0.4 N HNO_3 . Twenty litres of 25% v/v DBBP solvent in Shell* kerosene (flash point, 140° F) were prepared and equilibrated with 0.7 N HNO_3 before being added to the circuit. The stripping solution was made up from dilute sulphuric acid neutralized to pH 7.0 with gaseous ammonia. For pH control in stripping, a 2.0% ammonia solution was also prepared.

For each run the flowrates were held constant for the time required for a complete solvent cycle (6 hr). The aqueous phase from each settler and the extract and the recycle solvent were sampled before shutdown. A composite of the barren eluate for the whole series of runs was also collected and the phosphonate determined by an alkalimetric molybdate procedure after conversion to orthophosphate ion by a wet combustion procedure.

The pH in stripping was maintained by an automatic pH controller, feeding the weak ammonia solution to the first-stage mixer. In the



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subsequent mixer-settler stages the pH was controlled manually. The strip solution collected from each run was sent to precipitation and the barren solution was then recycled to stripping in the subsequent test, after being adjusted to pH 7.0.

Precipitation

The apparatus used for continuous precipitation was described in a previous report.⁽²⁾ Briefly, it consists of a 2000 ml beaker, provided with a heating mantle, pH electrodes and mechanical stirrer, to which are fed the strip solution and ammonia gas through separate flowmeters. The slurry overflows to a 1500 ml surge tank and thence to the filter. Where ammonia was diluted with nitrogen, the gases were metered separately before joining in a common tube leading to the bottom of the precipitation beaker.

Optimum conditions, as previously determined for continuous precipitation of ammonium diuranate (ADU) from amine sulphate strip solutions, ⁽²⁾ were used in the present tests, namely:

> pH: 7-7.5 Temperature: 30-60°C Retention time: 15-20 minutes Washing: By reslurrying

In this investigation, batch precipitation was compared with the continuous method. Also, the effects of temperature and gas dilution on the filtration, subsequent reduction and sintering of the ADU precipitate were studied.

The final ADU precipitates were submitted to the Research and Development Division of Eldorado Mining and Refining Ltd., Ottawa, for reduction and sintering tests.

RESULTS AND OBSERVATIONS

Extraction and Stripping

A series of nine runs was completed with eluates from the three sources. The results are presented in Table 1, and equilibrium data for three typical runs are shown graphically in Figures 2, 3 and 4.

Extraction Equilibria:

As the uranyl nitrate capacity of 25% DBBP was already known to be of the order of 90 g $U_3O_8/1$, the flowrates were adjusted accordingly. The effects of sulphate and nitrate concentrations in the eluate on uranium distribution (Co/Ca) were not clearly demonstrated but, within the range of concentrations covering possible variations in plant practice, it was established that barren eluates containing less than $0.1 \text{ g } U_3O_8/1$ were readily obtained in four countercurrent stages from a feed acidified to $0.4 \text{ N } \text{HNO}_3$.

Stripping Equilibria:

It was found that by adjusting the pH of the incoming solution to 6.0 and the pH of the first stage to 3.5, the pH in the 2nd, 3rd and 4th stages was automatically maintained in a range just below the point of uranium precipitation, where the stripping efficiency is highest. Under this condition, a minimum of three stages were required at high sulphate and low nitrate concentrations (Figure 3). Where the barren strip solution was recycled, a build-up of nitrate to 0.35 N NO3 occurred which increased the required number of stages to four when 8% (NH4)₂SO₄ solution was used and to five with 4% (NH4)₂SO₄ (Figure 4). In all

cases there is a well defined operating range which limits the uranium concentration to $45 \text{ g U}_3O_8/1$ in the strip solution and to $0.2 \text{ g U}_3O_8/1$ in the recycle solvent. This last value does not affect extraction efficiency, since it is still much below the equilibrium point of the fourth stage of extraction.

Phase Separation and Stage Efficiency:

Physically, the operation of the circuit was very satisfactory at the flowrates used. Although no quantitative data were obtained on settling rates, the use of 3 sq ft of settling area/gal/min of combined aqueous and organic phases was more than was necessary when operating with a continuous organic phase. Eluate retention time in each stage was approximately one minute, which gave almost 100% stage efficiency on the basis of the batch equilibrium data obtained previously with the same solutions.

Solvent Losses:

A composite barren eluate sample of the nine runs was analysed for solvent content and showed 0.10 g DBBP equivalent per litre (100 ppm). This represents a loss of 0.006 lb DBBP/lb U3O8 for an eluate containing 15 g U3O8/1. Soluble loss of DBBP has been reported previously(4) for a 2.5% solution in kerosene when in contact with acidic solutions; a value of 25 ppm was given. In the present investigation, where a 25% solution was used, the loss of 100 ppm does not seem unreasonably high.

TABLE I

Continuous Countercurrent Extraction and Stripping Tests. (25% DBBP)

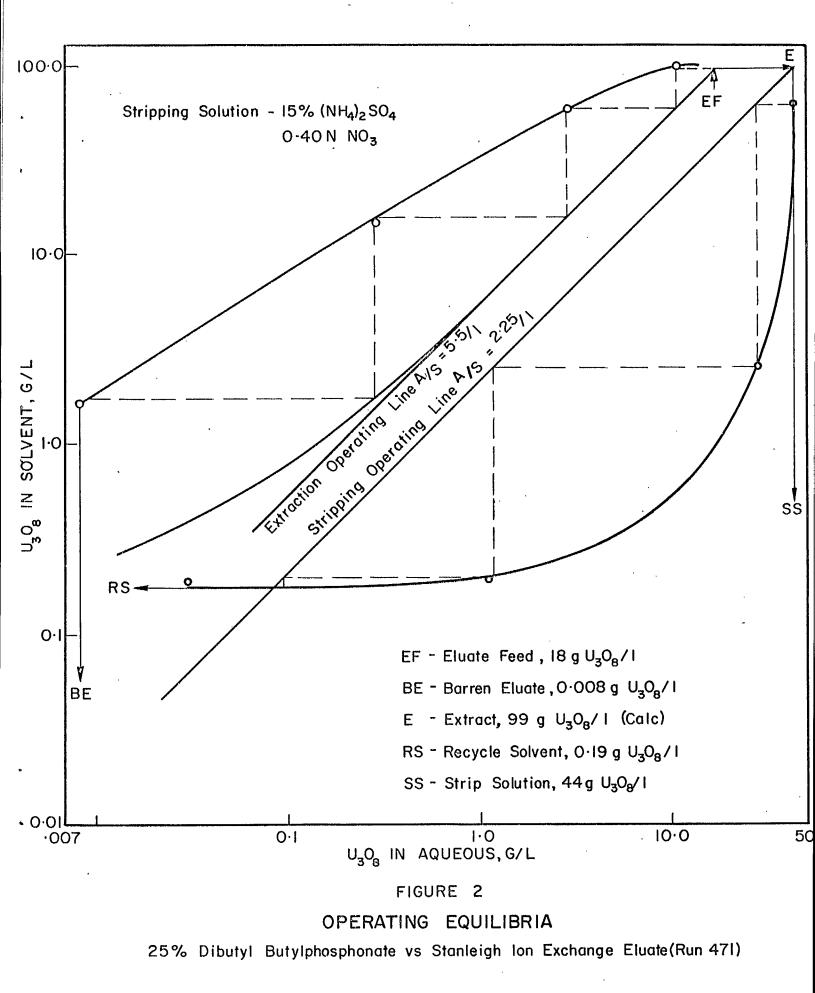
| RUN NO: | 469 | 470 | 471, | 472 | 473 | 474 | 475 | 476 | 477 |
|---|----------------|----------------|---------------|---------------------------------------|---------------|--------------|--------------|---------------|--------------|
| on Exchange Eluate | | Stan | leigh | · · · · · · · · · · · · · · · · · · · | Milliken Lake | | Consolida | ted Denison | |
| $U_{3}O_{8}, g/1$ | | | .19 | - , | 15.33 | | 14. 0. | | |
| HNO3 Normality Tot. NO3 " | | 1 | .70 | | 0.60 | | 1. | 20 | |
| ~ SO4, " | • | 0 | .64 | | 0.125 | | 0. | 405 | |
| tripping Solution | FRESH | RECYCLE | RECYCLE | RECYCLE | FRESH | FRESH | RECYCLE | RECYCLE | RECYCLE |
| $(NH_4)_2 SO_4 (g/1)$ | 193 | 183 | 152 | 164 | 193 | , 97 | 83 | 79 | 40 |
| NO3 (g/1) | nil 7.0 | 10.6 | 23.6 6.0 | 26,2 6.1 | nil 6.1 | nil 6.0 | 12.5 6.1 | 22.3 6.2 | 20.0 6.2 |
| pH | | 5.2 | | 0.1 | 0.1 | | | | |
| uration of Run, hours | 6.3 | 6,2 | 6,1 | 6.0 | . 6.0 | 6.0 | 5.6 | 5.8 | 6.2 |
| lowrates, ml/min | | | • | | | 1 | | | |
| Ion Exchange Eluate | 241 | 270 | 278 | 273 | 330 | 283 | 298 | 296 | 296 |
| Recycle Solvent | 62 | 59 | 54 | 50 | 54 | 55 | 50 | 49 | 49 |
| Stripping Solution | 122 | 117 | 121 | 120 | 103 | 109 | 99 | 98 | 109 |
| 308 Analyses, g/1 | | | | | | , | | | |
| Eluate: Stage 1 | 3,17 | 4.93 | 11.13 | 12.29 | 13,27 | 3,85 | 8,40 | 11.49 | 10.95 |
| 2 | 0.25 | 0.42 | 2.87 | 2,43 | 8.78 | 0.49 | 1.72 | 4.19 | 4.34 |
| . 3 | 0.015 0.002 | 0.026 0.003 | 0.29 | 0.14 0.005 | 4.09 0.30 | 0.041 | 0.14 | 0.58 0.046 | 0.96 0.10 |
| 4 | 0,002 | 0,000 | 0,008 | 0.005 | 0.50 | 0.000 | 0.012 | 0.040 | 0.10 |
| Strip: Stage 1 | 37.24 | 42.86 | 43.90 | 37.33 | 45.72 | 36.00 | 41.27 | 43.48 | 41.94 |
| 2 3 | 16.19 | 6.46 0.24 | 27.29 1.08 | 25,00 0,16 | 34.36 0.11 | 4.24 0.14 | 7.83 0.31 | 10.78 0.74 | 25,24 |
| 4 | 0.043 | 0.058 | 0.029 | 0.054 | 0.042 | 0,051 | 0.060 | 0.096 | 1.69 |
| Composite Barren Eluate | 0.005 | 0,002 | 0.006 | 0.006 | 0,21 | 0,017 | 0.010 | 0.027 | 0,064 |
| Extract | 74.76 | 83,85 | 92,61 | 94.31 | 92.01 | 64,76 | 80.42 | 86,73 | 80,81 |
| Composite Strip | 38.95 | 43,81 | 42,26 | 41.28 | 48.72 | 33.33 | 38.96 | 41.11 | 39.69 |
| Recycle Solvent | 0.14 | 0.20 | 0.19 | 0.25 | 0.23 | 0.20 | 0.20 | 0.24 | 0,55 |
| control pH in Stripping | | <i>,.</i> | | | | | | | |
| Stage 1 | 3,0 | 3.5 | 3.5 | 3.5 | 3.5 | 3.4 | 3.6 | 3,5 | 3.6 |
| 2 | 4.0 | 4.3 | 4.2 | 4.2 | 4.2 | 4.1 | 4.1 | 3.9 | 4.0 |
| 3 | 4.4 | . 5.1 | 4.9 | 5.1 | 5.5 | 4.5 | 4.5 | 4.5 | 4.3 |
| 4 | 7.2 | 7.2 | 5.3 | 5.9 | 6.4 | 5.0 | 5.1 | 5.5 | 5.3 |
| olumes, litres | | r | | | | | | | |
| Barren Eluate | 87 41 | 96 40 F | . 98 41 | 100 43 | 125 35,5 | 98 35,5 | 95 31.5 | 100 30 | 105 36 |
| Strip Solution | 41 | 40.5 | 41 | 40 | 33.5 | 35.5 | 51.5 | 50 | 50 |
| omposite Barren Eluate | | | | | 1 | | | | |
| ThO ₂ , $g/1$ | | | ,10 | | 0.10 | | | 07 | |
| (RE) ₂ O ₃ , " Fe, " | | | .01 .06 | | <0.01 0.05 | | | .01 .05 | |
| mmonia Consumption | | .* | | | | | * . | | |
| | 0,008 | , 0.011 | 0,007 | 0,018 | 0.005 | 0.009 | 0.013 | 0.008 | 0,021 |
| lb NH3/1b U3O8 | 0.008 | 0,011 | 0,007 | 0.010 | 0.005 | 0.009 | 0.015 | 0.000 | 0,021 |
| | | | | | | | | | • |

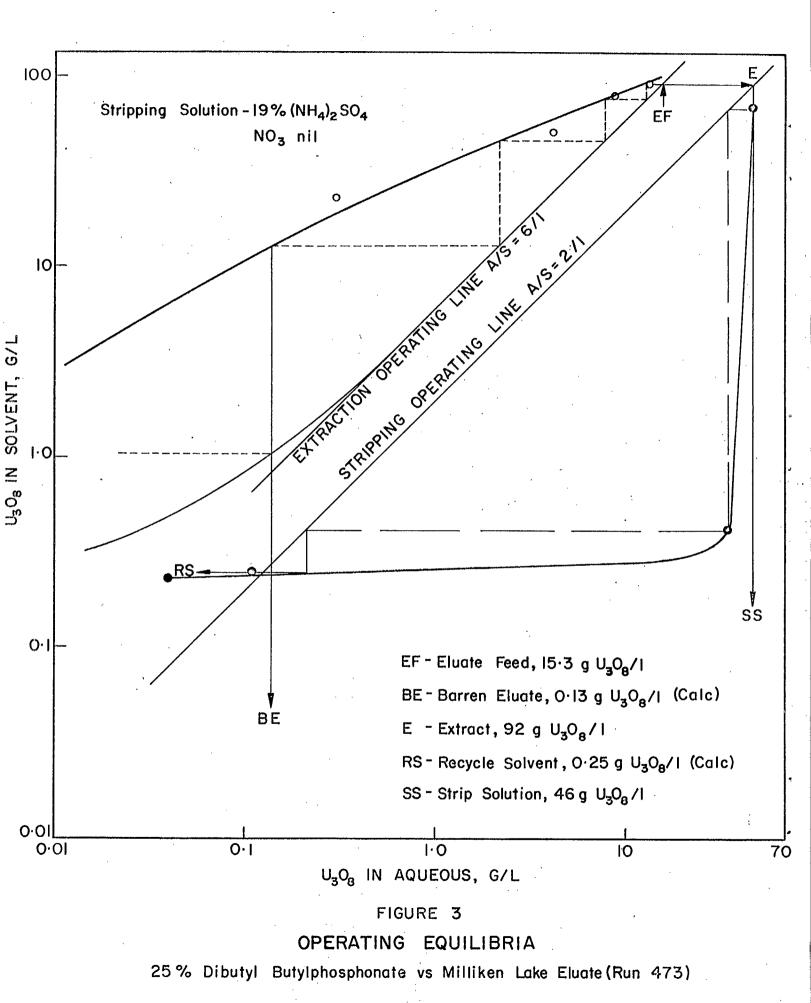
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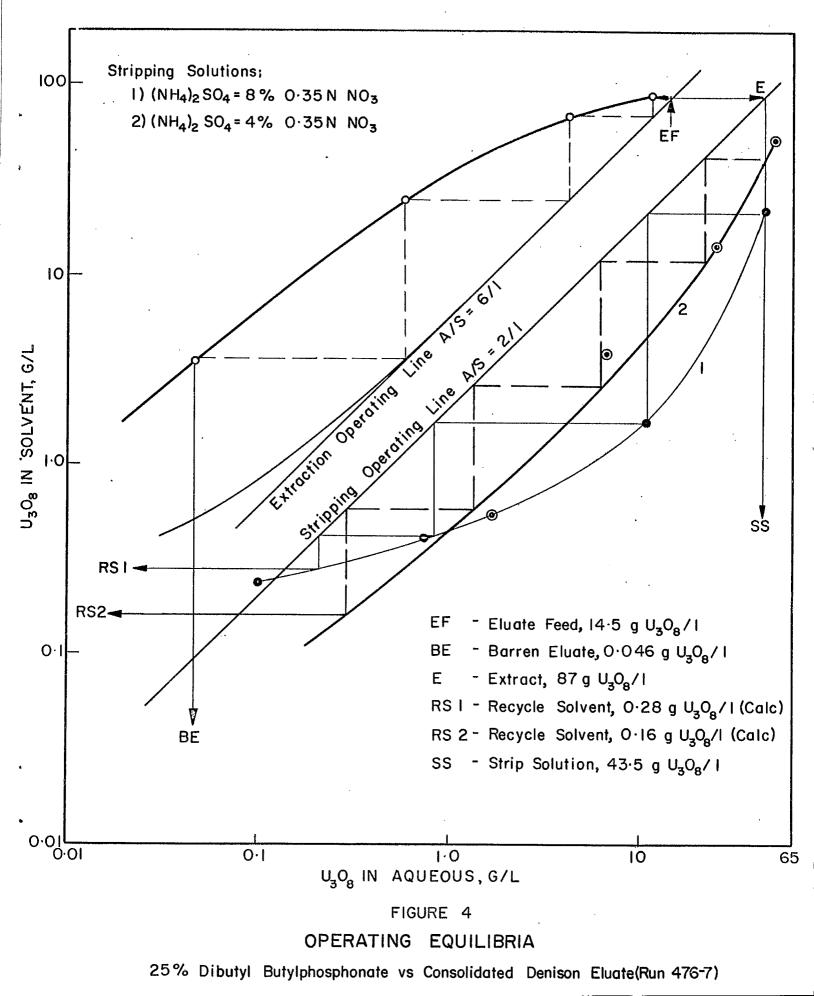
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Precipitation

A series of batch tests was done on Stanleigh strip solution, for comparison with the continuous procedure. The effects of temperature, pH and rate of precipitation on the sulphate content of the ADU were studied. The results are presented in Table 2.

TABLE 2

| | Batch Precipitation of ADU from S | tanleigh Strip Solution |
|------|-----------------------------------|-------------------------------|
| - | (Strip Solution: Initial pH, 3.3; | $U_{3O_8}, 37.14 \text{ g/1}$ |
| | | · · · |
| Test | Precipitation Conditions | ADU Draginitata Anglaria |

| No. | Preci | nitation Co | nditions | Precip | oitate Anal | lysis |
|-----|---------------|-------------|---------------------------------|--------------------------------------|------------------------|-----------|
| | Temp., °C. | Final pH | Time of ammonia add.(min) | U ₃ O ₈ , % | NH ₃ , % | SO4, % |
| 1 ` | 24 | 8.5 | 60 | 72.35 | 4.19 | 16.8 |
| 2 | 24 | 7.0 | 35 | 73.72 | . 4.08 | 17:19 |
| 3 | 40 | 7.0 | 190 | 74.07 | 4.34 | 15.87 |
| 4 | 40 | 7.0 | 35 | 72.30 | 4.84 | 16.41 |
| 5 | 40 | 8.5 | 65 | 66.12 | 4.45 | 16.41 |
| 6 | 70 | 7.0 | 35 | 75.0 | 4.48 | 14.07 |
| 7 | 70 | 7.7 | 220 | 76.87 | 3.75 | 14.13 |
| | | | | | | |

The UO_2 pellet produced from Test 6 ADU had a sintered density of 10.52.

In continuous precipitation tests, the effects were examined of temperature and of the dilution of ammonia with nitrogen on the sulphate content of the ADU and on the density of the UO_2 sintered pellets. The results are shown in Table 3.

The precipitate obtained from batch precipitation had a much higher sulphate content, which is chemically bound, probably as an ammonium uranyl sulphate double salt. Variations in the rate of ammonia additions did not affect the sulphate content, and precipitation at higher temperatures reduced it only slightly. Batch-precipitated ADU filtered much more rapidly than did the ADU precipitated continuously with 100% ammonia gas.

In continuous tests in which 100% ammonia gas was used, the chemical composition of the ADU was not affected by temperature. By diluting the ammonia with nitrogen, a coarser ADU was obtained which filtered well without the aid of flocculants. From tests 13 to 19 it will be seen that reproducible results can be obtained from continuous precipitation under the following conditions:

Retention time:10-15 minutespH:7.0-7.5Temperature:60°CAmmonia dilution:33% NH3 by vol. in air or nitrogen

Immediate filtration Washing by reslurrying.

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Continuous Precipitation of ADU with Gaseous Ammonia

| Test No. | Source of | | St | trip.Solutio | n | | Pre | cipitation Condi | tions | | | AD1 Precipitate | U e Analysis | | | UO2 Pellets |
|-------------|---------------|-----------------|------|--|-------------|-------------|--------------|----------------------------------|---------------------------|---------------|-----------|------------------------|-----------------|---------------|--------------|-----------------------|
| | Eluate | Vol., litres | pH | U ₃ O ₈ , g/1 | SO4, g/1 | NO3, g/1 | Temp., °C | Gaseous Ammonia, % by vol. | pH* of Barren Strip | U308, % | NH3, % | 50 ₄ , % | ThO2, ppm | RE2O3, ppm | B, ppm | Sintered . Density |
| 1 | Stanleigh | 15 | | 42.67 | | | 40 | 100 | 7.8 | 86.0 | . Z.88 | 3.90 | | | | 10.52 |
| 2 | 11 | 14 | ••• | ** | | | 60 | 11 | 8.8 | 85.7 | 2.99 | 5.40 | 28 | 26 | <i>.</i> 2.0 | 10.13 |
| 3 | 11 | 12 | | ** | | | 75 | н | 8.5 | 83.6 | 3.40 | 4.23 | | | | • |
| 4 | 21 | 30 | 3.0 | 35.71 | 115.5 | 18.1 | 70 | n | 8.6 | 80.4 | 2.98 | 5.80 | 56 | 13 | 0.3 | |
| 5 | " | 8 | | 42.65 | | | 11 | 33 . | 7.8 | 86.0 | 2.80 | 2.46 | | | | |
| 6 | 21 | 17 | | 11 | | • | IT . | 11 | 7.9 | 84.3 | 2.42 | 3.15 | | | | |
| 7 | | 10 | | | | | 11 | 11 | 7.8 | 87.7 | 2.62 | 1.47 | | | | |
| 8 . | Milliken Lake | 8 | 3.70 | 48.4 | 134.1 | 24.0 | 60 | 66 | 7.85 | 85.9 | 2.84 | 3.15 | | | | 10.53 |
| 9 | | 7 | 11 | ** | 11 | 11 | п | 50 | | 85.4 | 2.46 | 5.40 | 49 | <50 | | 10.50 |
| 10 | 11 | 5 | п | | 11 | 11 | 11 | 33 · | 7.80 | 79 . 7 | 3.71 | 8.58 | 56 | <50 | 4.0 | 10.53 |
| 11 | 11 | 9 | н | <i>t</i> 1 | ** | 12 | 21 | 25 | 7.75 | 87.7 | 2.88 | 3.45 | 53 | <50 | | |
| 12** | 11 | 2 | 11 | н. | 11 | 11 | 11 | 50 | 7.55 | 79.1 | 3.98 | 13.08** | 36 | 17 | <0.1 | |
| 13*** | Cons. Denison | 16 | 3.9 | 38.56 | · | | 60 | . 33 | 7.85 | 85.1 | 2.86 | 1.83 | 46 | < 5 | ċ 0.1 | 10.61 |
| · 14 | ** | 16 | 11 | ** | | | 11 | . II | 7.85 | 87.1 | 2.36 | 1.08 | | | | 10.58 |
| 15 | ** | 30 | 3.6 | 41.94 | · | | 11 | 11 | 7.90 | 86.7 | 2.80 | 1.68 | | | | 10.61 |
| 16 | 78 | 10 | ** | 40.38 | 45.21 | 39.4 | 12 | . 11 | 7.75 | 86.3 | 2.70 | 1.08 | | | | 10.76 |
| 17 | * * | 10 | 11 | . 11 | 11 | | 12 | . 11 | 7.70 | 90.0 | 2.27 | 1.29 | | | | 10.74 |
| 18 | 11 | 10 | 11 | 11 - | 11 | | f # | 11 | 7.70 | 85.7 | 2.98 | | | | | 10.62 |
| 19 | ** | 10.5 | 3.40 | 36.79 | | | 30 | ** | 7.45 | 86.0 | 2.74 | 1.80 | | | | 10.62 |

* Controlled pH during precipitation, 7.0-7.5. ** Batch precipitation. *** Tests 13 to 18: reproducibility tests with immediate filtration and washing by reslurrying.

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Purity of Product

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Dy

 Gd

In Table 4, the chemical and spectrographic analyses of the precipitates obtained by this process are compared with those of a refined ADU produced by Eldorado Mining and Refining Ltd.,⁽⁵⁾ Port Hope, Ontario. The values are comparable, but their significance in terms of nuclear grade purity can only be ascertained by actual neutron absorption tests as performed at Chalk River.

TABLE 4

Port Hope Refinery Element Source of Treated Eluate Specification Stanleigh Milliken Lake Cons. Denison of ADU Chemical % 85.7 85.4 85.0 86 U3O8, 2.66 NH3, 3.0 2.5 % -. 5.0 1.83 5.4 SO_4 , % 56 ThO₂, 16 46 50 ppm 13 17 RE2O3 ppm < 5 -0.3 < 0.1< 0.1 0.2 в, ppm Spectrographic(ppm, U basis) 1.0 < 0.1 Ag < 0.1 0.2 Cđ < 0.5 20 Co 10 ---< 5 10 10 \mathbf{Cr} 8 7 10 Cu <20 17 35 Fe 5 < 10< 1 Mn 1 < 106 Mo <15 2 15 Ni \mathbf{P} <40 10 40 15. 20 Si

<12

<0.06

<0.06

30

0.1

. 0.05

<10

0.2

0.21

Chemical and Spectrographic Analyses of Composite Ammonium Diuranate Precipitates

CONCLUSIONS

A refined uranium product has been obtained directly from current ion exchange nitrate eluates by extraction with dibutyl butylphosphonate in a continuous countercurrent mixer-settler unit. The ADU product obtained from the strip solution by continuous ammonia precipitation was of high purity and was used to produce uranium dioxide sintered pellets of densities greater than 10.5.

The process can be adapted to uranium leach plant flowsheets in which nitrate elution is used. Since the nitric acid concentration used in the liquid-liquid extraction is the same as is normally adopted for barren eluates recycled to ion exchange, nitric acid consumption will only be increased by the amount extracted by the DBBP solvent. In addition, the ammonia used for neutralizing this acid in stripping will represent an increase over that consumed in the precipitation of ADU. Solvent losses will also represent a small increase in reagent cost. All these items are shown in Table 5, which represents reagent costs over and above those of producing unrefined uranium "yellow cake" in the leach plant. The relatively high concentration of uranium in the solvent indicates that the circuit and the inventory of solvent will not be large.

TABLE 5

Costs of Reagents for Refining of Uranium from Ion Exchange Nitrate Eluates by the DBBP Process

| | Reagent | Reagent | | | | | |
|-----------------|-----------------|----------------------------|-------------------------------------|--|--|--|--|
| Reagent Used | Price, \$/lb | Consumption, lb/lb U3O8 | \$/1b U ₃ O ₈ | | | | |
| HNO3 (100%) | 0.07 | 0.037* | 0.0026 | | | | |
| NH3 | 0.045 | 0.01** | 0.0004 | | | | |
| $(NH_4)_2 SO_4$ | 0.02 | 1.25*** | 0.0250 | | | | |
| DBBP | 1.70 | 0.006 | 0.0102 | | | | |
| TOTAL | | | \$0.0382 | | | | |

* Calculated from the equivalent ammonia consumed.

** Averaged from Table 1.

*** Based on the use of 50 g (NH4)₂ SO₄/l, in a strip solution containing 40 g U₃O₈/l. The barren strip was not recycled.

REFERENCES

- A.J. Gilmore, V.M. McNamara, and R. Simard, "Tri-n-butyl Phosphate Extraction of Uranyl Nitrate from an Ion Exchange Mill Eluate for the Production of High Purity Uranium Dioxide". Mines Branch Investigation Report IR 60-84, Department of Mines and Technical Surveys, Ottawa, Canada, August 1960.
- A.J. Gilmore, V.M. McNamara, H.W. Parsons, and R. Simard, "Production of High-Purity Ammonium Diuranate by the Sulphuric Acid Elution-Amine Extraction Process". Mines Branch Investigation Report IR 60-95, Department of Mines and Technical Surveys, Ottawa, Canada, September 1960.
- T.H. Siddall, "Trialkyl Phosphates and Dialkyl Alkylphosphonates in Uranium and Thorium Extraction". Ind. Eng. Chem. <u>51</u>, 41-44, January 1959.
- C.A. Blake, D.E. Horner and J.M. Schmitt, "Synergistic Uranium Extractants: Combination of Neutral Organophosphorus Compounds with Dialkyl-phosphoric Acids". Report No. ORNL 2259, Oak Ridge National Laboratory, Oak Ridge, Tenn., 1959.
- 5. Eldorado Mining and Refining Limited, "Catalogue of Canadian Uranium Products for Research, Nuclear Power and Industry", Port Hope, Ontario, March 1959.

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