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OPERATIONAL REPORT ON THE PUREX PLANT THORIUM PROCESS TEST

> JP Duckworth DG Harlow CC Herrington

March 15, 1965

PREPARED FOR THE U.S. ATOMIC ENERGY COMMISSION UNDER CONTRACT AT(45-1) 2130

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By

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Purex Operation CHEMICAL PROCESSING DEPARTMENT

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Operated for the Atomic Energy Commission By the General Electric Company Under Contract # AT (45-1)-1350

HANFORD ATOMIC PRODUCTS OPERATION Richland, Washington

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INTRODUCTION

Long-range planning for Hanford production is based on the efficient use of the Purex Plant as a multipurpose facility. One of the variety of products proposed is U-233, obtained from the irradiation of thorium oxide. Since there was a lack of sufficient data to completely define a processing run of thorium in the Purex Plant, a test run was proposed using five tons of thoria to better define general technology, develop an operable flowsheet, and demonstrate the flexibility of the various equipment systems.

To process such a small amount of feed and still produce a U-233 product with a usable specification, it is necessary to completely flush the plant of U-238 and alter the normal process routes so that the U-233 can be separated from thorium and both materials purified for shipment.

The operations and changes required for such a run were considered desirable on a test basis in order to obtain experience and information on the projected "campaign" mode of operation. Also, detailed evaluation of the efficiency of types and methods of flushing in all sections of the plant was considered necessary to aid future planning.

Combined, all of the information generated from such a test would enable a reasonable estimate of plant turn-a-round time to be made which would be invaluable in future planning and scheduling activities for the Purex Plant and Hanford.



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SUMMARY

The Purex Plant Thorium Process Test was successfully completed February 10, 1965, after 25 days of processing. Three special products were loaded out: 2886 grams of U-233, 7660 pounds of thorium, and 4 grams of protactinium-233. The U-233 was 95.5% pure, which exceeded the expected purity of 90%.

Extensive flushing of the plant was required before and after the run to ensure product purity. A total of 450,000 gallons of solution was used to carry out 185 individual flushes described in 93 procedures. The presence of a small amount of uranium bearing siliceous deposits in the dissolver pot and auxiliary equipment extended the time estimated for flushing this system by approximately fifty-five days. The problem was resolved by replacing the dissolver pot and downdrart tower. Seventy days were devoted to flushing the head end section of the plant and 37 days were used for the solvent extraction section. Eight days of terminal flushing followed the test.

Thoria targets were charged to the dissolver on January 11, 1965, and the solvent extraction operation began January 25, 1965. Essential material costs for the test included \$35,800 for flushing and \$11,000 for processing with \$5,200 charged to solvent extraction processing. Six hundred and twenty thousand gallons of unconcentrated waste were generated by the run.

Since development flowsheets were used to process thorium and uranium-233, all existing operational and control specifications and procedures had to be reviewed



SUMMARY (Con't)

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and revised to ensure operating safety within the limits defined by Research and Engineering. Valuable process control experience was obtained from the run and associated procedure preparation. Considerable laboratory experience was also gained during the test with more than 15,000 analytical determinations being made on 4,000 samples.

An unusually heavy burden was imposed on the canyon crane operations in order to reroute, maintain and replace remote equipment and jumpers. A total of 230 jumper changes and nine major equipment lifts were necessary to complete the test.

The Thorium Test was completed with relatively few problems. Besides the detection . of trapped solids in the dissolver off-gas system, the only other significant problem was the rapid decrease in the solvent quality during the thorium decontamination run.

CONCLUSIONS

The Thorium Process Test demonstrated that irradiated thoria fuel can be processed efficiently in the Purex Plant. All three products, U-233, thorium and protactinium were successfully recovered and loaded out. No serious problems were encountered that cannot be resolved by flowsheet, operating or equipment changes for future thorium processing. Except for the detection of solids in the dissolving equipment and the rapid solvent degradation during thorium processing, the various aspects of the test proceeded as originally predicted for the process test conditions.



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CONCLUSIONS (Con't)

Estimates of plant turn-a-round time have been better defined as a result of evaluation of the equipment changes and flushing operations carried out for the thorium test. For future multipurpose operation, the following nominal turn-a-round estimates* are made:

Plant turn-a-round from normal processing to thorium processing - 2 weeks. Plant turn-a-round from thorium processing to normal processing - 1 week. Plant turn-a-round from normal processing to E metal processing - 1 week.

Plant turn-a-round from E metal processing to normal processing - 1 week. These values are based on average operating times for equipment changes and flushes. Where the crane operations are limiting, an average of eight remote jumpers can be installed per shift in a single canyon opening, or an average of four remote jumpers can be installed per shift in multiple canyon openings and the remote replacement of a major equipment piece averages 1 to 2 shifts after the jumpers are removed.

For empty-out operations, five days are required to purge the plant so that less than 150 units of each product are retained in either the head end or solvent extraction sections.

To further reduce the residual product hold up in the head end section to less than 50 grams, an estimate of eight days is required for the clean out of a standard Purex dissolver and four days for the new annular dissolver using a boiling 12<u>M</u> nitric acid, 0.025<u>M</u> fluoride, and 0.05<u>M</u> ANN solution. For tanks or equipment with efficient agitation such as in D and E cell, a maximum of four days is required for clean out.

*Estimates can be very sensitive to product specifications.

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CONCLUSIONS (Con't)

Complete clean out of the solvent extraction equipment to less than 50 grams can be accomplished with three cycles of flushes which can be completed in nine days. A cycle is composed of a flushing series of 20% nitric, 3% nitric acid, water, 5% caustic-5% tartrate, and water, which generates approximately 50,000 gallons of concentrated waste solutions per cycle for a total of 150,000 gallons for the clean out.

For future multipurpose operation, use of the canyon cranes will have to be phased directly into the processing sequence for canyon routing changes in addition to the normal crane function of remote maintenance and equipment replacement. Since processing steps will be dependent on crane work to this unusual degree, the ease of remote handling of canyon equipment should be included in the primary criteria for future design. Also, future equipment design should be reviewed to minimize traps or dead spots which could hold up product containing solution.

The successful application of closeup remote TV for the inspection of the bottom of the C3 dissolver proved its effectiveness as an inspection tool. The scrap detected on the bottom of the dissolver indicates that more positive control of metal transfers needs to be invoked when annular dissolvers are installed, to minimize the possibility of fuel jamming in the annular dissolver or sections of the bottom being plugged. In addition, such material is undesirable should a weight input measurement system be instituted in the Purex Plant.

DISCUSSION

Brief reports on the information gathered during the Thorium Process Test are presented to support the conclusions of the test.

Thorium Run Plans

The first document for a Purex Plant Thorium Process Test was issued in July, 1964. In the five months that followed, the plans for the plant scale process test took form.

<u>Feasibility and Equipment Needs</u> Thorium processing in the Purex Plant was judged to be feasible after a detailed study by the CPD Research and Engineering Operation. However, certain aspects of the flowsheet and the performance of plant scale equipment were not completely defined with the available data. To further evaluate the flowsheets and operational problems which might arise, a process test was approved using five to six tons of thorium.

<u>Planning and Scheduling</u> Preliminary flowsheets for the process test had to be defined early so that the following activities could commence:

- 1. Ordering remote canyon jumpers for new routings.
- 2. Revising critical mass control specifications to include Th and U-233.
- 3. Preparing flush specifications and procedures.
- 4. Issuing operating procedures.
- 5. Writing operational control specifications.
- 6. Developing analytical methods and sample schedules.

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Thorium Run Plans (Con't)

When these items had been, prepared in draft form, a Hazards Review was conducted by a CPD task force. Planning was, therefore, a joint effort of all CPD and other HAPO components, as several interrelated items were passed between groups until a satisfactory solution was found.

The original schedule called for the C3 dissolver to be removed from normal uranium processing in time to allow fifteen days for flushing prior to charging with the irradiated thoria target elements. This flush of the dissolver and protactinium recovery vessels was to remove residual normal uranium and plutonium. Dissolution and protactinium recovery would then take thirty-three days, during which time the normal uranium processing would be completed and the solvent extraction equipment stripped and flushed (see Figure II). The schedule allowed eleven days from the completion of the normal uranium run in which to flush the solvent extraction equipment and make the necessary routing changes preparatory to running thoria.

The total solvent extraction time, including partitioning, U-233 purification and thorium purification was scheduled to require twelve days.

Post-run flushes and restoration of plant equipment to normal was planned to take sixteen days.

This schedule was not met due to the difficulty in removing residual uranium and plutonium from the original dissolver and the necessity of developing equipment routings to avoid uranium contamination even after a new dissolver and repaired dissolver tower were installed. However, as can be seen from Figure II, once the



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Thorium Run Plans (Con't)

thoria was charged the original schedule was either attained or surpassed in terms of elapsed time.

The dissolution flowsheet, with centrifugation of the Routing and Equipment Changes coating waste and protactinium recovery, required the installation of four new routes dissolver (C3) to centrifuge feed tank (E3), centrifuge catch tank (E5) to coating waste tank (D2), centrifuge slurry tank (E1) to dissolver (C3) and centrifuge catch tank (E5) to concentrated backeycle waste tank (JI) so that the installed centrifuges and storage vessels in E Cell could be used. A route from dissolver (C3) to the acid recovery system 1WF tank (F12) (see Table I) was provided to transfer the reflux condensate from the dissolver off-gas tower to the acid recovery facilities. This route was designed to allow the reflux condensate to be directed either to the dissolver as is normal or to F12. A hold-up pot was built into the jumper connecting the dissolver off-gas knock-out pot and the dissolver. A jet built into the hold-up pot could be activated to direct the reflux condensate to F12. This route was intended for use only during the acid removal step but it was used during dissolution to avoid introducing contamination from the dissolver tower. A special monitor was installed on the dissolver coil to detect any coil failure resulting from the corrosive solutions used in the thorium dissolution. The off-gas monitors normally used to detect radioiodine emission from the C3 dissolver were specially calibrated to alarm should radioruthenium evolution occur and not be captured in the off-gas tower and filters.

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Thorium Run Plans (Con't)

A shielded cask was provided and the fission product load-out facilities were used for the load-out of the protactinium recovery product.

The small amount of feed to be processed made it advisable to employ a 5% TBP solvent extraction flowsheet similar to ones previously developed at ORNL and SR for thorium -U-233 partitioning the U-233 purification in the Neptunium Recovery and Final Plutonium cycle equipment. However, the higher capacity flowsheet was also tested during the thorium purification run in the large solvent extraction equipment. The use of these two distinct flowsheets in the process test made extensive equipment and routing changes necessary (Table I).

The U-233 recovery required three routes -- solvent makeup tank (RIA) to 1EXF tank (J3), J3 to HAO tank (J2) and J2 to the No. 1 Solvent System (G Cell) header -- to modify the J2 and J3 tanks for makeup, storage, and feed of the 5% TEP solvent to the solvent extraction columns. Special routes were also required to transfer the thorium-bearing waste to storage vessels, J-22 aqueous waste (3AW) to 1WF tank (F12) and F12 to 1WW neutralization tank (F16). A route was required for the U-233 product from the partition cycle to the final uranium purification cycle, J-23 product (3EN) to 2AF tank (J5). The L Cell Package equipment (stripper T-L3, concentrator E-L4 and product receiver tank TK-L6), which was used to concentrate the U-233 product, was not geometrically favorable for all concentrations of U-233, so special safety circuits and instruments were provided as a backup for the batch size control system of critical mass control. Modifications were required in the Product Removal Room so that the U-233 load-out into the special M-102 "bird cage" carriers could be performed. The thorium purification run required canyon jumpers and cold side piping so that three non-routine

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streams (HAX-HNO3 to the HA (H2) column and HSIS-HNO3 and HSS-Butt to the HS (H3) column) could be added to the solvent extraction columns. A special route, 2DF tank (K1) to concentrated backcycle waste or 3WB tank (J1), was provided so the thorium product could be "spun" back to the feed tank for additional solvent extraction treatment. Three jumpers were required to route the final thorium product to the load-out facilities (K1 to 203 load-out line). The product was loaded into a trailer and transported to the 200-W Area for storage in the specially re-activated WR vault.

<u>Procedure Preparation</u> A great deal of time and effort was spent in preparing specifications due to the magnitude and complexity of the Thorium Process Test. Separations Process Engineering with the assistance of Purex Process Control performed the bulk of this "paperwork." The specifications and procedures required the review and approval of several organizations to assure safe operation.

The desired product purity, coupled with the purity of the thoria target elements before processing, dictated the solvent extraction cleanup necessary and the amount of contamination from normal processing that could be allowed to remain within the processing equipment. When the processing flowsheets were finalized, the flushing and operating procedures were written.

A sample schedule was prepared by Separations Process Engineering, after which Purex Analytical Control, assisted by the Process Chemistry group developed and



Thorium Run Plans (Con't)

adapted analytical procedures with capabilities for process control.

The processing of a fissionable isotope other than Pu-239 and natural uranium in the Purex Plant made revision of the critical mass control specifications necessary.

New essential material and chemical hazards specifications were required, since the Thorium Process Test used some chemicals not previously used at Purex, e.g., phosphoric acid, manganous nitrate and potassium fluoride.

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Flushing requirements were particularly stringent to assure product purity for the relatively small amount of material to be processed. Forty-five separate flush procedures were written and used for the dissolver and related equipment before the vessels were judged clean enough to charge the target elements. In many cases, new routings had to be provided so the flushing could proceed with a minimum of time and material expended.

A typical listing of the magnitude and variety of specifications and procedures follows:

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DEGLASSIFIED THORIUM PROCESS TEST SPECIFICATIONS AND PROCEDURES

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- A. Specifications
 - 1. Product Specifications
 - 2. Flushing Specifications
 - 3. Feed Specifications
 - 4. Essential Material Specifications
 - 5. Critical Mass Control Specifications
 - 6. Chemical Hazards Control Specifications
 - 7. Chemical Flowsheet Specifications
 - 8. Operational Control Specifications
 - 9. Sample Schedule

B. Procedures

- 1. Flushing Procedures
 - a. Dissolver and Pa Recovery Equipment
 - b. Partitioning and U-233 Purification Equipment
 - c. U-233 Concentration and Load-out Facilities
 - d. Thorium Purification Equipment
 - e. Thorium Load-out Facilities
 - f. Miscellaneous Routings
 - g. Post Thorium Run Flushes of All the Above Equipment

2. Operating Procedures

- a. Use of UO3 Acid During Thorium Run
- b. Aqueous Makeup Flushes and Processing
- c. Coating Removal and Centrifugation
- d. Thoria Dissolution and Denitration



Thorium Run Plans (Con't)

- e. Protactinium Recovery and Solvent Extraction Feed Adjustment
- f. Solvent Extraction U-233
- g. U-233 Load-out
- h. Solvent Extraction Thorium
- i. Thorium Load-out
- j. Solvent Treatment
- k. Waste Concentration and Nitric Acid Recovery

<u>Hazards Review</u> A team was appointed from within CPD to review the plans for processing the thoria in order to evaluate the hazards involved in the proposed methods of operation.

The task force concluded that the greatest areas for concern were in avoiding radioruthenium volatilization with possible environmental contamination and critical mass control, especially in the non-geometrically safe U-233 concentration equipment.

Special procedures and controls were deemed adequate for the original test run but the processing of large quantities of thoria would require redesign of some systems and revaluation of the hazards involved.

For further details, see Reference 12.

Pre-Test Plant Clean Out

The target isotopic purity of the U-233 was set at 90%. Because the total U-233 to



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Pre-Test Plant Clean Out (Con't)

be processed was so small (\sim 3000 grams) the estimated allowable U-238 to be left in the plant equipment was 150 grams. The Plant was divided into three sections for flushing purposes, the recovered acid system, the dissolver and E Cell equipment, and the solvent extraction system. The target for each of the sections was a retention of 50 grams of U-238. These goals were met, as the isotopic purity of the U-233 product was 95.5% (see flush summary below).

<u>Recovered Acid System</u> Flushing of the recovered acid system was relatively easy. The source of uranium in the recovered acid $(U\dot{O}_3 \text{ recovered acid at 0.07 pounds uranium per gallon of acid}) was valved off on October 22. Dilution by process turnover reduced the Purex recovered acid system uranium content to between one and four pounds of uranium per million gallons by November 18. This was well within the specification for use in the thorium processing.$

UO₃ recovered acid was used during the month and a half of normal processing after the above line was blanked, because neither Purex nor the UO₃ Plant has capacity to store the amount of acid generated with the uranium product (0.5 pounds HNO_3 per pound of U). Purex used the UO₃ acid directly in the dissolvers after a program was started to insure a uniform acidity. The UO₃ Plant set up special procedures to assure a constant UO₃ recovered acid concentration and shipped only from a tank which was known to meet a specification of 49 to 52.5% HNO₃. As a double check, the UO₃ recovered acid was sampled and checked at Purex before use. The acid used in the dissolver must be within the limits of 49 to 52.5 percent to make sure the dissolving solution will not become acid deficient under normal operating control limits.

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<u>Dissolver and E Cell Equipment</u> Dissolver C3 was chosen as the dissolver to be used for thoria dissolving. The last charge of uranium was made on 4 - 12 shift, October 31. Following two normal cuts, the heel was removed by two full cuts and a heel cut. The actual flushing started on day shift November 3, when 1600 gallons of 57 percent nitric acid were added to a 400-gallon heel cut. This flush, No. 1 on Table III, was boiled for 24 hours under reflux. The flush removed 26 pounds of uranium, indicating that the heel cut procedure does a good job of removing the readily-dissolvable uranium. To keep from diluting the dissolved feed in the feed storage tank (D4), the acid from C3 was jetted to dissolver B3 by special jumpers at the dissolver rinse tank (D1). The acid was used in B3 for a normal dissolution. Two more 57 percent acid flushes were made on C3 until it was found that the heel had been sufficiently removed so that acid flush solutions could be sent to the acid recovery system without excessive loss of products. A route was then established to the acid recovery system feed tank, 1WF tank (F12).

Operation of E Cell for fission product recovery was suspended on November 12. The first flush sent from dissolver (C3) to centrifuge feed tank (E3) to start flushing the E Cell equipment was flush No. 4, a 20 percent HNO flush (see Table III). The flushing of E Cell continued with the dissolver flushing until the task was completed. The head end flushes are summarised in Table II and shown individually in Table III. Until the rest of the plant was shut down, all acid flushes went to the lWF tank (F12) and about 75 percent of the acid was recovered.

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Pre-Test Plant Clean-Out (Con't)

After 27 flushes totaling 64,630 gallons, it was found that the amount of uranium removed from dissolver C3 had leveled off at a value greater than 10 grams per hour, too great to accept as a final clean-out. At this point, the flushing conditions were intensified, the acid - ANN/fluoride (2:1) solutions were boiled and in some cases the fluoride concentrations were raised. The intensified flushes took three general courses: high nitric (57%) low fluoride (.025 M) with 2:1 ANN at boiling; low nitric (28%) medium fluoride (.5 M) with 2:1 ANN added after heating at 80° C; and low nitric (20%), high fluoride (1.0 M) with 2:1 ANN added after heating at 90° C. Silica-base' solids were suspected to be causing the high dissolution rates because the plutonium to uranium ratio was about four times the normal ratio. It was assumed that the dissolved iron in the flush solutions came from the dissolver coils because the coil surface was at a higher temperature than other surfaces. The average corrosion rate for mild flushes was 8.5 mils per month. For more intense flushes the rate was 15 mils per month. At the maximum concentration of 2.0 M fluoride the corrosion rate was 70 mils per month for the four-hour flush period (ANN was added after the four-hour heating period on a 2:1 mole ratio). An estimated 3.5 mils were removed from the coil of the original dissolver. The original wall thickness of the coil was 150 mils.

On December 17, a TV inspection of the dissolver pot showed the dissolver was extremely clean of product solids except for miscellaneous scrap from a broken charging bucket and 100 Area experimental equipment (See Photos 1 and 2). At this point, a new dissolver was installed on the basis of the uncertainty of the dissolver content. The new dissolver was in place on December 21.

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Pre-Test Plant Clean-Out (Con^t)

After the new dissolver was installed, flushes intended to remove welding and fabricating debris showed small amounts of uranium in each flush (See Table III). It appeared to be coming from the dissolver tower via the reflux return line and possibly from the expansion bellows in the vapor line from the dissolver to the tower. Seventeen flushes of the new dissolver and old tower removed about eight pounds of uranium. The Pu/U ratio of the product removed from the tower was still in the range of 2200 grams/ton. The rate of dissolution was still too high, at 10 to 100 grams per hour, so the tower was removed on January 3.

A 10-inch pipe jumper was fabricated to by-pass the tower knock-out pot and ammonia scrubber. Three boilups with flush solutions were made. The uranium dissolution rate decreased but ruthenium contamination appeared to be a problem. The off-gas monitors which had been modified to measure Ru-103 and -106, alarmed during all of the The vapor-handling capacity of the off-gas system was also a problem. A boilups. dissolver tower which had been removed from dissolver B3, in February, 1964, was repaired in T Plant on an accelerated work schedule and installed on C3. The tower had been scheduled to be used when the new annular dissolvers were installed. An estimated 3 to 4 weeks was required to repair the leaking tubes and mock it up on a normal schedule; however by working around the clock on an "all-out" schedule, it was completed in four days. After installation, this tower was also found to have some uranium contamination. The uranium/plutonium ratio was the same as the original tower. It was decided to install a plug-blank in the reflux return line and jet all condensed reflux to the lWF tank (F12) through a jumper route already installed for use during acid removal.



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Pre-Test Plant Clean Out (Con't)

When the plant completed normal processing in early December, all the flushing was routed from E Cell to coating waste tank (D2). After neutralizing, this material was jetted to non-boiling waste storage.

Flushing of E Cell equipment was continued in conjunction with dissolver flushing. All the flushes were sent through E Cell, whichever route of disposal was taken: to acid recovery system, 1WF tank (F12) or to coating removal waste tank (D2). G-E4, a B Plant-type centrifuge, was new and unused but was included in the flush schedule as a backup for centrifuge G-E2. This proved to be fortunate as the diverter valve on G-E2 failed during the last portion of the protactinium run and G-E4 was used to complete the run.

<u>Solvent Extraction Equipment</u> The solvent extraction system shutdown was started on December 5. Cold feed was processed to remove gamma activity from the system so the final stripping would not produce a gamma burst. The cold feed also displaced plutonium from the system so the backcycle system contents could be reduced in volume after shutdown and stored in the HAF tank (H1).

The final strip and displacement of columns was the normal procedure used for an inventory clean out. After the plant was shut down, a 10 percent nitric acid and water flush (pre-flush) was put through all solvent extraction systems which were to be used in the thorium - U-233 run (See Figure III and Table IV). This flush was intended to remove the gross quantities of uranium, plutonium, and neptunium so that subsequent decontamination flushes could be thrown away. The pre-flush was collected in 3WF tank (FlO) and boiled down batchwise in the backcycle concentrator (H4). A

sample of each batch was used to monitor product content. This pre-flush solution was combined with the backcycle waste, the complete batch was reduced in volume by boiling off nitric acid, then stored in HAF tank (H1). The pre-flush product content was very low at 52 pounds uranium and 136 grams plutonium, indicating the normal shutdown clean-out is very effective. The flushing with 10 percent nitric acid by cascading forward consumed a large volume of flush and was slow because the routes available were limited by the flush requirements and other shutdown activities. For instance, early flushes from the J Cell Package could not be routed through L Cell because their high uranium concentration would unnecessarily contaminate the uranium-free equipment. Therefore, L Cell and J Cell Package were flushed separately requiring extra coordination to schedule both flushes through the 2AF tank (J5). Another delaying circumstance was that the lEX column (J6) empty-out route shared a nozzle with the 2DF tank (K1) to 3WF tank (F10) route, and therefore, required extra coordination to schedule the right route to be in place.

The decontamination flushes following the pre-flush consisted of a series: 20 percent nitric acid, water, 10 percent caustic-2 percent tartaric acid, water, 30 percent nitric acid-5 percent oxalic acid. Two cycles of these flushes were generally used. A summary of the flushes is shown on Table IV. The total product removed by the decontamination flushes was 48 pounds of uranium, 31 grams plutonium and 60 grams of neptunium. The points of greatest hold up and resistance to flushing were found to be as predicted: 3AF tank (J21), the ICU concentrator (EJ8), and the 2DF tank (K1). Additional flushing was required to reduce these vessels to desired uranium limits.

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Pre-Test Plant Clean Out (Con't)

The following flush summary outlines the flush limits by processing groups and compares the limits with actual results:

PRE-THORIUM FLUSHING SUMMARY

First Cycle System

Limits

0.0012 grams plutonium per gallon in J6, J4, J5

0.0001 pounds uranium per gallon in Jl, H2, H3, J3, J6, J7, J8, Kl

Level Attained

0.0007 grams plutonium per gallon in J6, J4, J5

0.00004 pounds uranium per gallon in Jl through Kl

Material Removed

Plutonium: 10 grams, Uranium: 20 pounds

Neptunium Recovery System and Second Plutonium System

Limits

Less than 60 grams uranium in J21, J22 and J23

Less than 100 grams plutonium in J21, J22, J23 J5, L1 and L2

Level Attained

Approximately 40 grams uranium in J21, J22 and J23

Less than 1 gram plutonium in J21 through L2

Material Removed

Plutonium: less than 25 grams, Uranium: 20 pounds Neptunium: 25 grams.

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Pre-Test Plant Clean Out (Con't)

L Cell Package and Product Removal Vessels (L3, L4, L6, L9, L13)

Limits

Less than 0.12 grams plutonium per gallon

Level Attained

0.075 to 0.12 grams plutonium per gallon

Material Removed

Approximately 50 grams plutonium

Processing Operation

The experience of processing thoria in the Purex Plant was entirely new, and as such it was given close attention in order to gain the maximum value from the process test.

<u>Coating Removal</u> On January 11, 1965, 9489 pounds of thoria were charged to the dissolver. The physical equipment comprising the Purex dissolving system was in its normal configuration with the exception of some jet routes.

The thoria was charged in the normal manner and the aluminum jackets were removed following the Standard Operating Procedures. Before removing the coating waste, the dissolver was digested for six hours with sparging to maximize removal of aluminum which may have been covered or coated with the thoria fines. The off-gas from the coating removal step was scrubbed for ammonia removal and routed to the stack, bypassing the backup facility.

The coating waste solution plus two 1000-gallon rinses were transferred to E Cell

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Processing Operation (Con't)

for recovery of entrained thoria. In the E Cell the coating waste and rinses were centrifuged and the resulting cake washed with caustic to remove the aluminum solids. Then the cake was slurried out of the centrifuge bowl to slurry tank El where it was digested in a nitric acid, fluoride, ANN solution to dissolve the thoria. Samples of this solution indicated a pickup of two percent of the thorium charge. It was planned to recycle this solution to the dissolver, but because of fifty percent contamination of the U-233 with U-238, the solution was discarded to underground storage.

Before the thoria dissolution was begun, the acid reflux recycle route to the dissolver from the dissolver tower and knock-out pot was blanked. This was done to prevent the residual U-238, which was trapped in the tower and bellows connectors, from contaminating the dissolver cuts.

Thoria Dissolution and Denitration Dissolving was started on January 12, 1965, using 12.2M HNO3, 0.025M KF and 0.05M ANN solution. The off-gas from the dissolving operation was handled in the normal manner. The reflux from the tower and knock-out pot was continually jetted to the 1WF tank (F12) for acid recovery. Batch additions of fresh 57% HNO3 acid were made to the dissolver to maintain a volume of 1700 to 2200 gallons.

The dissolution rate of the thorium was monitored by observing the buildup of the dissolution solution specific gravity. The first cut was cooled for sampling after 25 hours of boiling. At this time, a new sampler jumper was installed in the



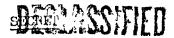
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Processing Operation (Con't)

dissolver. The jumper dip leg was shortened by seven inches to allow sampling of solution without plugging the intake leg.

The dissolver solution was then heated back to boiling and acid boil-off begun. This was accomplished by adding 500 gallons of water, concentrating to 1700 gallons and repeating this for a total of 21 hours. The first cut was cooled, allowed to settle with minimum air on the sparger for one hour and then jetted to centrifuge feed tank (E3). The total time required to dissolve and adjust the acid for this cut was 55 hours. At the end of the first cut a new thermohm, 13 inches off the bottom of the tank, was installed to give better control of solution temperature. The second cut was begun immediately following the completion of crane work and continued until January 17, 1965. At this time acid adjustment began. Samples of the first cut indicated a HNO3/Th ratio of 1.95 rather than the desired 1.5 so the first cut was brought back to the dissolver and both the first and second cuts were acid adjusted simultaneously. The feed acid adjustment procedure was altered by using steam through the sparger lines and in the dissolver heating coil instead of through the coil alone. With this additional heating capacity and with the combined volume of two cuts, the thorium concentration was increased more easily without compromising the minimum volume limitation. In approximately 24 hours, the acid adjustment of the combined cuts was completed and they were moved to Tank E3. Then a third cut was begun. The third cut continued for a total boiling time of 47 hours ending January 21. The fourth cut continued for a total boiling time of 35 hours ending January 23. The fifth cut continued for a total boiling time of $36\frac{1}{2}$ hours ending January 24 (see Figure I and Table V).

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Processing Operation (Con't)

Starting January 25, cuts 3, 4, and 5 were combined in C3 dissolver and acid adjusted in the same manner as cuts 1 and 2. At this time jumpers were changed to route the reflux from the tower and knock-out pot to 1WW tank (F7) rather than the 1WF tank (F12) to avoid diluting the 3AF from the U-233 run. Acid adjustment was completed on January 26. The combination of cuts 3, 4, and 5 was transferred to E Cell and C dissolver was held pending return of routings available to dispose of final flushing. The complete cleanout of the dissolver is described in the following section.

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<u>Protactinium Recovery and Loadout</u> - Following acid boil-off from the combination of the first and second cuts, the dissolver solution was transferred to E Cell for protactinium recovery. The protactinium run began January 19 and continued until January 21, for a total of 47 hours of processing time.

The protactinium run began by adjusting the dissolver feed solution to 1.0 M HNO₃ and 0.1 M sulfamic acid. Then manganous nitrate and potassium permanganate were added to the solution to form manganese dioxide to scavenge the protactinium 233. All butt solutions were added to centrifuge catch tank (E5) and the cake carrying the Pa-233 was slurried to slurry tank (E1). In Tank El the cake was dissolved in a nitric acid sugar solution and the resulting solution was recentrifuged to remove the highly radioactive barium sulfate. During centrifuging the diverter, which directs the centrifugate to either Tank E3 or Tank E5, malfunctioned, allowing approximately 50 percent of the Pa solution to mix with the solvent extraction feed in Tank E5.

A second MnO_2 strike was made to remove the thorium and U-233 from the remaining Pa-233. For this strike Centrifuge G-E⁴ was used at 1750 rpm. The final product



Processing Operation (Con't)

was then loaded out into a 400 gallon cask in the Fission Product Loadout Building (Table VI). Radiation levels during loadout were 50 mr/hr exposure to personnel. No difficulty was encountered in the loadout.

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A second protactinium run was made on the accumulated thorium purification cycle waste (HAW). The volume of this waste was 6000 gallons, three times greater than in the first Pa strike. Pre-strike samples showed 28 units of protactinium were available but after the strike less than ten units could be detected in the slurry catch tank (E1). Besides this discrepancy in Pa units, the feed scavenging step concentrated the fission products in the precipitate, making the probability of meeting the product contamination specifications remote. So, rather than chance contaminating the first strike product, the second strike was discarded to boiling waste storage.

<u>Uranium-233 Recovery and Loadout</u> - The U-233 recovery portion of the thorium run utilized the following equipment and routes: Centrifuge catch tank (E5) to 3WB tank (J1) to 3AF tank (J21) to 3A column (J22) to 3B column (J23) to 2AF tank (J5) to 2A column (L1) to 2B column (L2) to 2BP stripper (L3) to 2BP concentrator (L4) to product receiver tank (L6) to sampler tank (L9) to loadout (equipment nomenclature joined by to indicates a route prepared especially for this run).

The feed solution was adjusted to -0.2 pounds/gallon HNO (acid deficient) in 3WB tank (J1) (this signifies that the solution lacks two tenths pound of nitric acid per gallon of solution from being in stoichiometric balance). On January 25,

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Processing Operation (Con't)

J Cell Package and L Cell were started up. In the 3A column (J22), the feed was contacted with a five percent solution of TBP in diluent, the 3AX extractant stream. The scrub stream, 3AS, contained iron, ANN, and phosphoric acid. The iron (ferrous ion) was to insure U/Pu decontamination. The ANN was for salting the U and the phosphoric acid to insure U/protactinium decontamination.

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In the 3A (J22) column the U-233 favored the organic phase and the thorium remained in the aqueous. The aqueous waste (3AT) containing the thorium was collected in F Cell (Tank Fl2 and Tank Fl6). The U-233 was stripped from the organic with 0.05 \underline{M} HNO₃ in the 3B (J23) column. The aqueous was then continuously jetted via the 3BN jet system to the 2AF (J5) tank. The stripped organic was returned to the organic storage tank J3.

The U-233 in tank J5 was continuously butted with HNO_3 and Fe^{++} and contacted in the 2A (L1) column with 5 percent TBP, stripped in the 2B (L2) column with 0.01 <u>M</u> HNO_3 and concentrated in the L Cell Package. Little difficulty was encountered with equipment operation (see Figure IV).

With the critical mass limit placed on the U-233 content in the L Cell Package (1300 units maximum), it was necessary to control the buildup of product in the "package" by specific gravity instrumentation and material balance calculations. There were instances of conflicting information between the indicated and calculated accumulation. During one "package" empty-out, the steam jet malfunctioned, causing concern for the mass limit. Because of these circumstances, the product

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was transferred to the load-out tanks before it reached the desired concentration, never exceeding sixty-nine percent of the limit. The first batch from the L Cell Package was moved to the sampling tank (L9) on January 26, after 20 hours of accumulation, at a concentration of 13.6 g/1. This transfer and subsequent transfers were accomplished by cutting off the stripping stream (2BX) and product stream (2BP) to the 2B (L2) column and moving the 2BP stripper (L3) and concentrator (L4) contents to the product receiver tank (L6) and then recharging L3 and L4 with fresh 1 \underline{M} HNO₃. The L3 and L4 contents were then brought to boiling and the 2BX and 2BP returned to normal rates.

The first and second load-out batches were very dilute (see Table VII) because of extra caution used in the critical limit control point. For these two batches, 2BP flow was used for control (see Sketch I). The analyses necessary to operate solvent extraction and control critical mass were often delayed. This was due to the overload of analyses requested from the Laboratory with routines, re-runs and resamples, and the length of time required per sample, such as 105 minutes for thorium by the ion exchange method.

The second U-233 batch was transferred to sample tank (L9) at 0030 on January 27 (after 10 hours of collection) at 10.8 g/l total U.

The third U-233 batch was transferred to Tank L9 at 1330 on January 27 (after 13 hours of accumulation) at a concentration of 31.7 g/l total U.

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Processing Operation (Con't)

The fourth U-233 batch was transferred to Tank L6 at end of stripping L Cell after shutting down the solvent extraction system. Its concentration was 30 g/l. The product was loaded out via load-out tank L13 into one-gallon plastic jugs. A total of 48 jugs were loaded out, including the four product batches and $l\frac{1}{2}$ flush batches. The jugs were contained in M-102 "bird cage" containers.

Final flushes containing $\angle 5$ g/l U-233 and ~ 0.04 g/l Pu were loaded into 5-gallon plastic jugs for storage.

<u>Thorium Recovery and Loadout</u> - Following the U-233 recovery run, the 5 percent TBP solution in Tanks J2 and J3 were butted to 30 percent TBP. The organic header system was restored to normal the IBXF jumper installed. The organic in J2 and J3 was transferred to the No. 1 organic treatment system (G Cell) through the IBX (J6) column and IC (J7) column. G Cell was "spun" to clean up the solvent.

From January 27, the end of the U-233 run, until January 29, the beginning of the thorium run, routings were restored to conditions necessary for thorium processing; eleven jumpers were involved. Also, during this period, the feed for the thorium decontamination run was transferred from F Cell to E Cell from 1WF tank (F12) to 1WW neutralization tank (F16) to centrifuge feed tank (E3) and through the centrifuge to remove any traces of solids picked up from the F Cell vessels, then adjusted to feed specifications in the centrifuge catch tank (E5) and the 3WB tank (J1). During this period, a leak was discovered in the left hand remotable tube bundle of the 1WW concentrator (E-F6). The tube bundle was not changed at this time, but merely isolated, requiring eight hours of crane time.

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Processing Operation (Con't)

Difficulty was encountered in the makeup and analysis of HSS butt solution. This solution was initially planned to be 1.76 \underline{M} Fe⁺⁺ and 1.75 \underline{M} PO₄-3. Due to the low solubility of the PO₄-3, the makeup was changed to one-half strength.

The thorium run utilized the following equipment and routes: 3WB tank (J1) to HA column (H2) to HS column (H3) to 1BXF tank (J3) to 1BX column (J6) to 1C column (J7) where partitioning took place. The thorium went to the 1CU concentrator (J8) to 2DF tank (K1) to 3WB tank (J1) for Phase II, and to 203-A loading station for load-out to a trailer. (Equipment nomenclature joined by to indicates a route prepared especially for this run.)

The HA-HS column battery was used for decontamination from fission products. These columns were operated as normal HA-HS columns except that an additional scrub stream, the HSIS (fresh HNO₃) was added to the HSIS distributor and (recovered HNO₃) -HAX-HNO₃ was added to the normal HA nitrite distributor at the bottom of the HA column. The thorium and U-233 in the organic phase overflowed the HS column to the lEXF tank (J3), was pumped through the lEX column, and the thorium was partitioned by stripping out of the organic in the lC column. The thorium was then concentrated in the lCU (J8) concentrator which overflowed to the 2DF (K1) tank. Trace amounts of uranium remained with the organic and flowed to G Cell (see Figure V). The initial throughput of the acid-deficient thorium feed began on January 29, at 1530 and ended January 30, on 4-12 shift. It was difficult to establish equilibrium operating conditions because of the erratic feed control in the 3WB system and the small amount of feed. Higher-than-expected losses were experienced in the HAW stream.



Processing Operation (Con't)

Phase II of the thorium decontamination run consisted of recycling the thorium product from the 2DF (K1) tank to the 3WB tank (J1) and running this as HAF. This phase began on January 30, on 4-12 shift. The difference between Phase I and Phase II was that the HAF was acidic in Phase II. After approximately 25 hours of running on Phase II, the HA column (H2) pulser failed. The process was shut down and the HA column contents displaced and the pulser changed out. During shutting down, replacing the pulser and starting up there was considerable loss of thorium. This loss was both to the cell sumps due to equipment maintenance and to the HAW waste due to the upsetting of the process by stopping and starting with such a small volume of feed. The 1CU (J8) concentrator was emptied to the 2DF (K1) tank and the 1WW (E-F6) concentrator was emptied to 1WW tank (F7).

While shut down, it was decided to switch back to an acid deficient flow sheet for startup, in an effort to obtain better decontamination of the thorium. The process was restarted up on February 2, on 8-4 shift after 64 hour shutdown to replace the HA pulser. Also, the flow control problem on the HAF (3WB) system was corrected by blanking off the pump recycle line. This was a prototype pump recycle jumper with a magnetic flowmeter. The run was terminated on February 4, on 4-12 shift after approximately 50 hours of Phase II operation for a total operation time Phase I and II of 98 hours.

At shutdown, the columns were stripped. The total thorium load was accumulated in the 1CU (J8) concentrator and then transferred as one 2000-gallon batch to the 2DF (K1) tank. The concentrator was then rinsed with one water flush to tank K1



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ending up with a total volume of 2500 gallons. This material was then loaded out to a tank truck via the 203-A loading station and transferred to 200-W Area for storage in the WR vault. Five subsequent flushes of the 1CU (J8) concentrator and 2DF (K1) tank were also loaded out to a truck and moved to 200-W. A total of 7600 gallons were loaded out (see Table VIII).

Near the end of the thorium recovery cycle the No. 1 Solvent System quality dropped to well below normal. The organic Pu retention (measure of quality) increased up to 100 times its normal value (see Figure VI): After attempting to clean up the solvent by recycling through G Cell, with no apparent improvement, G Cell was shut down and the vessels flushed. Solid manganese dioxide was found in samples throughout the system, indicating that the acid recycle rate was not great enough to kill the entrained solids. All of G Cell was flushed with oxalic-nitric solution to dissolve the manganese dioxide. This procedure required five days.

Post-Thorium Cleanout and Turn-Around

The specifications for thorium and uranium-233 impurity in normal uranium and plutonium products required that the dissolver and the solvent extraction equipment be thoroughly flushed (see flush summary below).

<u>Dissolver Cleanout</u> - A series of dissolver flushes were made immediately following the four product cuts and the fifth heel cut (see Table V). Number 6 and 7 (flushes) utilized nitric acid-fluoride solution identical to that used during earlier cuts.

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Post Thorium Clean Out and Turn-A-Round (Con't)

Flushes 8, 9 and 10 comprised a sequence of 57% nitric acid flushes, the last of which digested for approximately 50 hours. A final water rinse followed the above flushes. Normal dissolving operations began on February 5.

Total material removed from the dissolver amounted to approximately 250 pounds of thorium and 100 grams U-233.

<u>Solvent Extraction Flushes</u> In general, flushing of the extraction equipment required much less time following the run than before the run. This was primarily due to the thoroughness of the initial flushes, the relatively short period between flushes, and the higher limit of contamination allowed after the run.

The flushing goals and actual results are presented in the flush summary below. Product levels were calculated from vessel volumes and sample data taken during flushing. It is important to note that the absolute quantity of product in a given piece of equipment could be more than the figures given. The product levels given are derived from a sizeable tabulation of sample data and are considered to be the best available estimates (see Table IX).

First cycle columns HA (H2) and HS (H3) were flushed separately from the remainder of the first cycle equipment, to avoid fission product contamination. Earlier equipment flushing results indicated that fission products were retained longer by the columns than the product materials. The product levels attained in columns H2 and H3 after three 10% nitric acid flushes were low enough to begin normal uranium processing.

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Post Thorium Clean Out and Turn-A-Round (Con't)

First cycle equipment, columns 1EX (J6) and 1ES (J4), the 1EXF tank (J3) and 2AF tank (J5), was given a separate flush. Samples taken in J5 (the last vessel in the flow scheme of the flush) reached low product levels rapidly. First cycle vessels, column 1C (J7), 1CU concentrator (J8) and 2DF tank (K1), were also flushed as a unit. U-233 levels in the terminal vessel of this flush were very low (see the flush summary below).

Five 10% nitric flushes were applied to the J Cell Package neptunium equipment. Sample data taken during the flush indicated a greater than normal product holdup in the J Cell Package feed tank. Product levels attained are shown in the summary below.

The second cycle plutonium equipment was decontaminated quite thoroughly after four 10% nitric flushes. Product levels dropped to less than one gram of U-233 and less than 13 grams of thorium per thousand gallons.

The plutonium stripper-concentrator pair (L3 and L4) was extensively flushed along with product removal equipment to insure a low degree of contamination from U-233 and thorium.

POST THORIUM FLUSHING SUMMARY SOLVENT EXTRACTION

First Decontamination Cycle (T-H2 through E-J8)

Limits

Less than 20 grams thorium in J3 and J6

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Post Thorium Clean Out and Turn A-Round (Con't)

Less than 300 grams thorium in H2, H3, J7, J8

Less than 20 grams U-233 in H2, H3, J3, J6, J7, J8

Level Attained

Approximately 130 grams thorium in H2 and H3 after flushing

Approximately 250 grams thorium in H2 through J8 during cold operability run

Less than 20 grams U-233 in H2, H3, J3, J6, J7, J8

Material Removed

Thorium: 14 pounds, U-233: less than 5 grams

J Cell Package - Neptunium Recovery Equipment

Limits

Less than 300 grams thorium

Less than 20 grams U-233

Level Attained

Approximately 500 grams thorium

Approximately 1 gram U-233

Material Removed

Thorium: 1 pound, U-233: 5 grams

Second Cycle Plutonium System

Limits

Less than 10 grams thorium

Less than 10 grams U-233

Level Attained

Less than 1 gram thorium in J5, L1, L2, L3, L4

Less than 1 gram U-233 in J5, L1, L2, L3, L4

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Post Thorium Clean Out and Turn-A-Round (Con't)

Material Removed

Thorium: Less than 25 grams, U-233: Less than 2 grams (J5, L1, L2)

75-100 grams (L3, L4)

<u>Solvent Treatment</u> The No. 1 Solvent System furnished solvent for the first cycle thorium decontamination. The solvent and solvent equipment were essentially unchanged for the thorium decontamination phase of the run.

Analytical data relating solvent quality were obtained from the usual solvent plutonium retention analysis (Figure VI). It was noted during the thorium decontamination run that the solvent quality decreased (high plutonium retention) during the time when acid deficient thorium feed was used and increased during the time when acid feed was used. Examination of the system indicated an entrainment problem in the solvent purification equipment. A solid formed and normally confined to the solvent system feed tank (G1), was found to be distributed throughout all of the solvent system equipment.

At the end of the thorium run, the No. 1 Solvent System (G cell) equipment was flushed with 18,500 gallons of 5% oxalic acid, 3% nitric acid flush solution in seven flush throughputs. The oxalic-nitric flush dissolved the manganese dioxide solids. The system was returned to spinning status and the solvent was found not to be degraded.



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Associated Operations and Incidents

Canyon equipment changes during the thorium test run required concentrated activity by the canyon cranes to meet the processing schedule (Figure I). Nine major lifts of dissolver equipment were required during replacement of the C3 dissolver and tower in preparation for thoria charging. Some twenty-six new jumpers were installed, which in turn required removal of fifty-three existing jumpers.

Twelve temporary flex routings were installed during the run, for flushing and waste processing. Equipment failures such as concentrator tube bundle leaks, and HA column (H2) pulser failure, contributed emergency crane work and some delay in processing.

Timely maintenance on the cranes made it possible at times to utilize both cranes effectively, doing cell work at opposite ends of the canyon. The most critical and closely scheduled period for crane work was in early February when the test run was completed and it was necessary to restore the plant for normal processing. Three shifts of the jumper work, needed to restore fission product processing capability, were delayed until after startup to make time for higher priority work and are not shown on the schedule (Figure VII). In general, the jobs completed which were not directly related to the test run are typical of a Purex shutdown although the number of equipment failures was above average. Three concentrator tube bundles failed and were replaced, in addition to the HA column pulser, 1WF pump and a dozen failed jumpers.

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Associated Operations and Incidents (Con't)

Efforts were successful in removing failed canyon equipment such as the F6 concentrator, F15 tank and assorted failed jumpers during the period. This is normal canyon activity during a shutdown.

While not apparent from the schedule, the crane work performed in support of the test run exceeded reasonable expectations in every phase of the run. The skill and experience of the crane operators was one of the intangible but very heavy plus factors in successful completion of the run.

During the dissolver flushing stage of the test, it became necessary to view the inside of the dissolver. This was undertaken with a remote television camera in a stainless steel carrier which was lowered into the dissolver to determine if undissolved material remained. By remote cable connection, the picture was relayed to the Operating Gallery for general viewing (see Photos I and II). Some undissolved material was in evidence under the grating. An unexpected collection of stainless steel scrap, observed in the bottom of the dissolver, was presumed to have collected over the years from foreign material in slug buckets dumped during charging. ,

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LIST OF FIGURES AND TABLES

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Figure I Figure II Figure IV Figure V Figure VI Figure VII	Solvent Extraction Flush Details Sample Analysis of U-233 Product Streams Sample Analysis of Thorium Product Streams
Table I Table II Table III Table IV Table V Table VI Table VIII Table IX Table X	Pre-Test Dissolver Flushing Details Summary of Solvent Extraction Flushes Thorium Dissolving and Denitration Protactinium Recovery U-233 Material Balance Thorium Material Balance
Sketch I	Thorium Process Flow Sketch
Photo I Photo II Photo III	TV View of C3 Dissolver Scrap on Dissolver Grid Reclaimed Tower at T Plant

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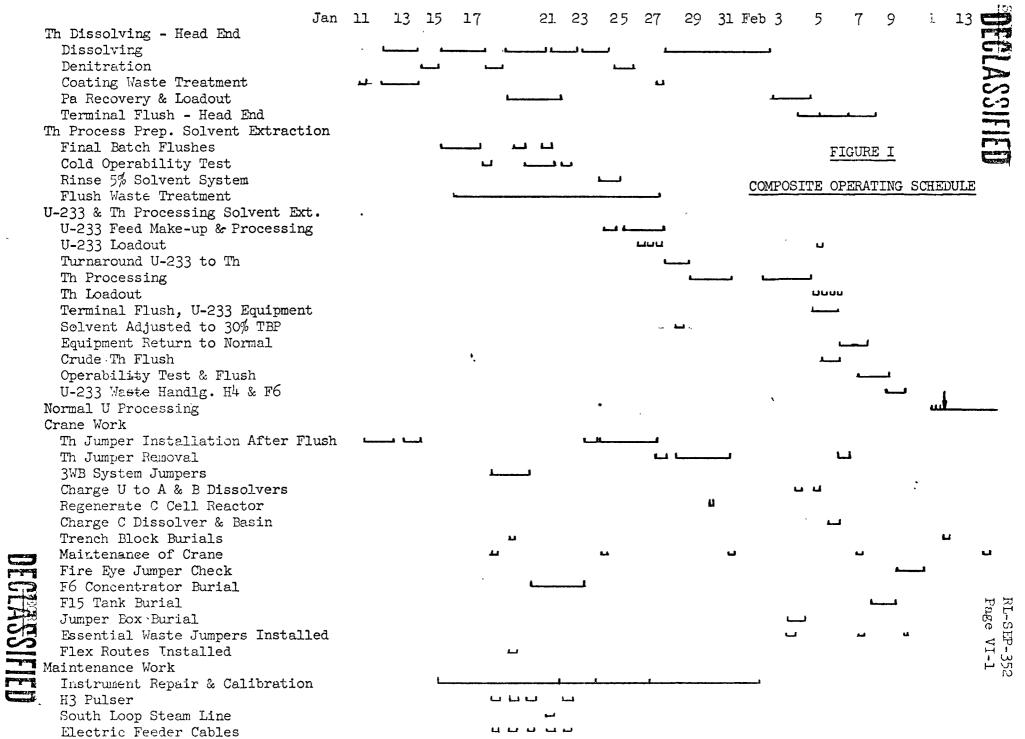
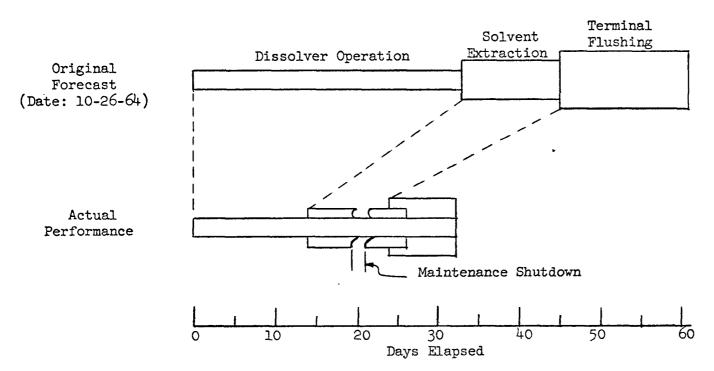


FIGURE II

ORIGINAL	OPERATING	FORECAST
	VERSUS	
ACTUA	L PERFORM	ANCE



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FIGURE III

	FIGURE III SOLVENT EXTRACTION FLUSH DETAILS	MENTA 00
Equipment Systems Flushed	$\frac{12-7}{12-8} \frac{12-9}{12-10} \frac{12-11}{12-12} \frac{12-13}{12-14} \frac{12-15}{12-15} \frac{12-16}{12-17} \frac{12-18}{12-19} \frac{12-20}{12-20} \frac{12-2}{12-20} \frac{12-2}{12-20}$	
L Cell Package Deep Flush	<u> </u>	-
J5-L1-I2		-1
J Cell Package		<u>5</u> 1
J1-H2-H3-J3 J6-J7-J8-K1		21
Kl-Jl Kl-Flo		긕
J7-J8-K1 K1-J1		
J6-J4-J2-J5		
Backcycle Spin H4 MT out	┝ ╌ ┥ ┝ ╌ ┥	
Gl Flush G7 Flush		
Rl Flush		RL-SEP-352 Page VI-3
	1. Water 4. 5% Oxalic-30% Nitric 2. 20% HNO3 5. 10% Caustic-2% Tartaric 3. 10% HNO3 6. 30% HNO3	- 352 I - 3

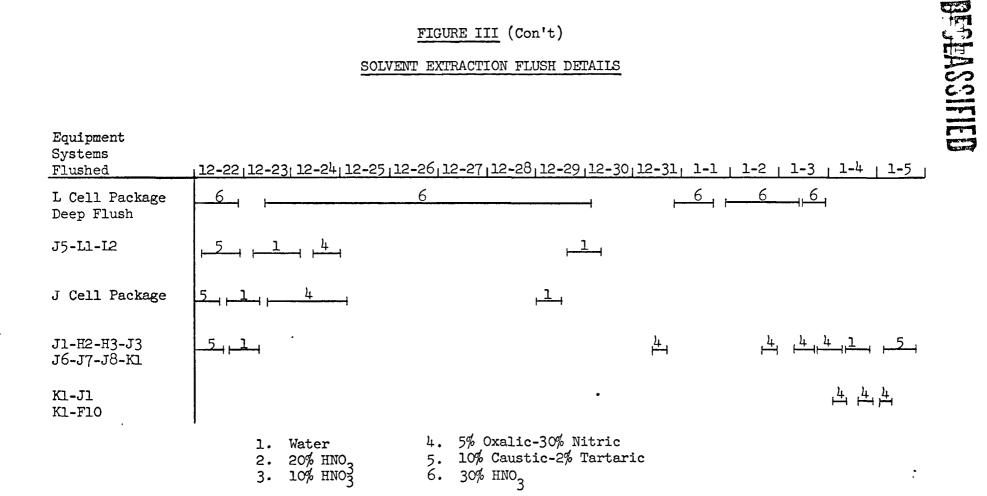
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FIGURE III (Con't)

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SOLVENT EXTRACTION FLUSH DETAILS



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FIGURE III (Con't)

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SOLVENT EXTRACTION FLUSH DETAILS

Equipment Systems Flushed	1-6 1-7 1-8 1-9 1-10 1-11 1-12
L Cell Package Deep Flush	
J5-L1-12	
J Cell Package	
J1-H2-H3-J3 J6-J7-J8-K1	
KI-JI KI-FIO	
,	1. Water 4. 5% Oxalic-30% Nitric 2. 20% HNO3 5. 10% Caustic-2% Tartaric 3. 10% HNO3 6. 30% Nitric

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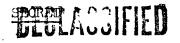
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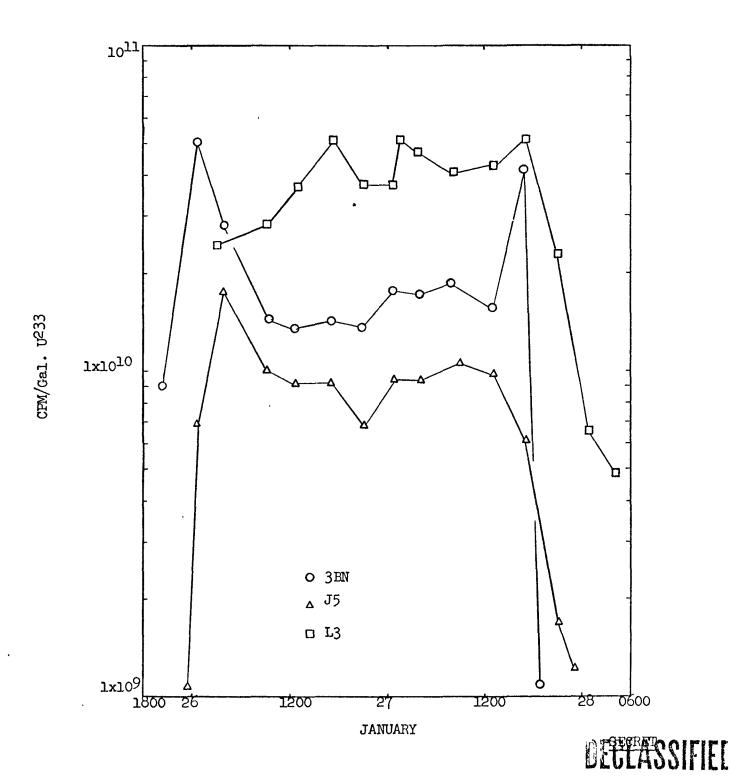


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FIGURE IV

SAMPLE ANALYSES, OF U233 PRODUCT STREAMS



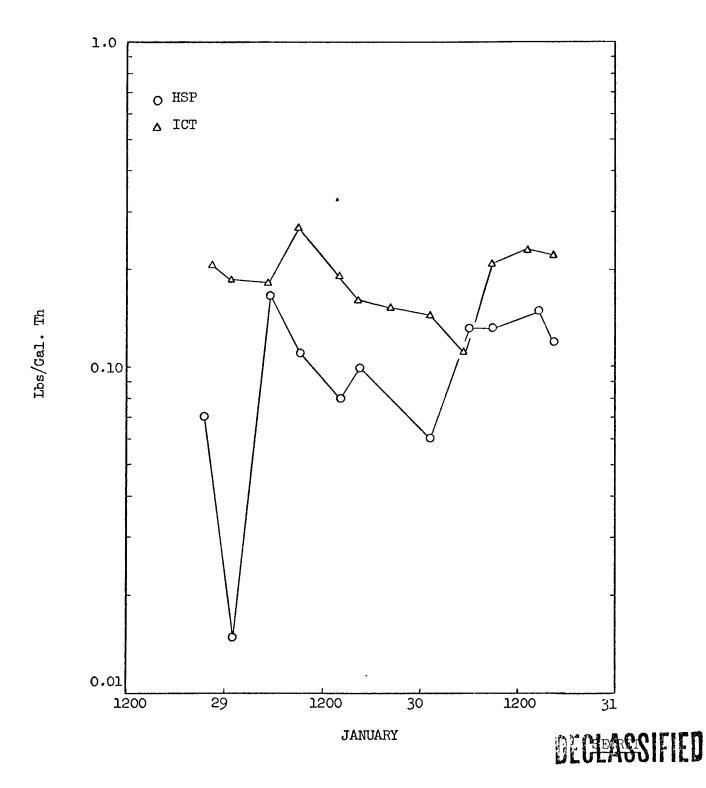


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FIGURE V

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SAMPLE ANALYSES OF THORIUM PRODUCT STREAMS

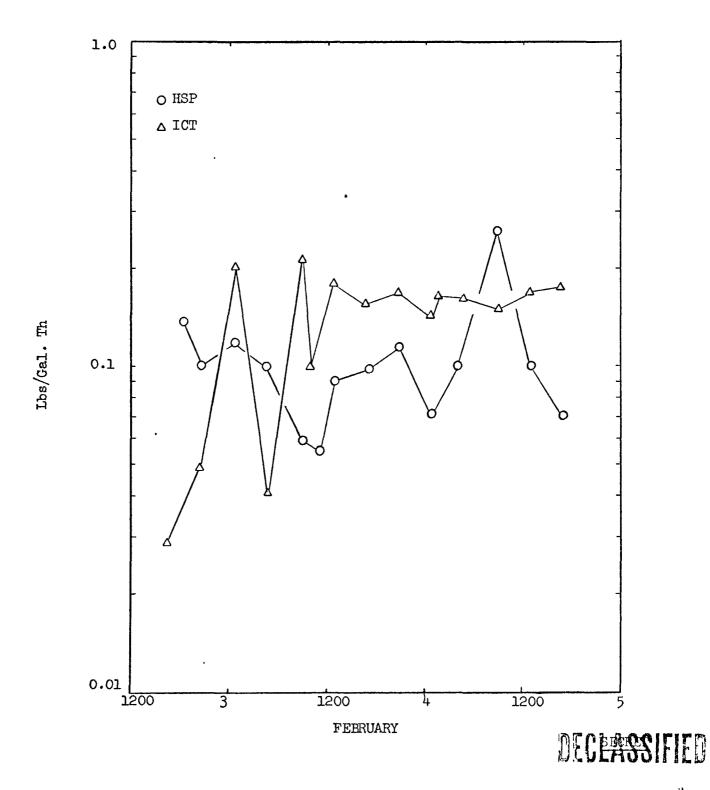




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FIGURE V (Con't)

SAMPLE ANALYSES OF THORIUM PRODUCT STREAMS



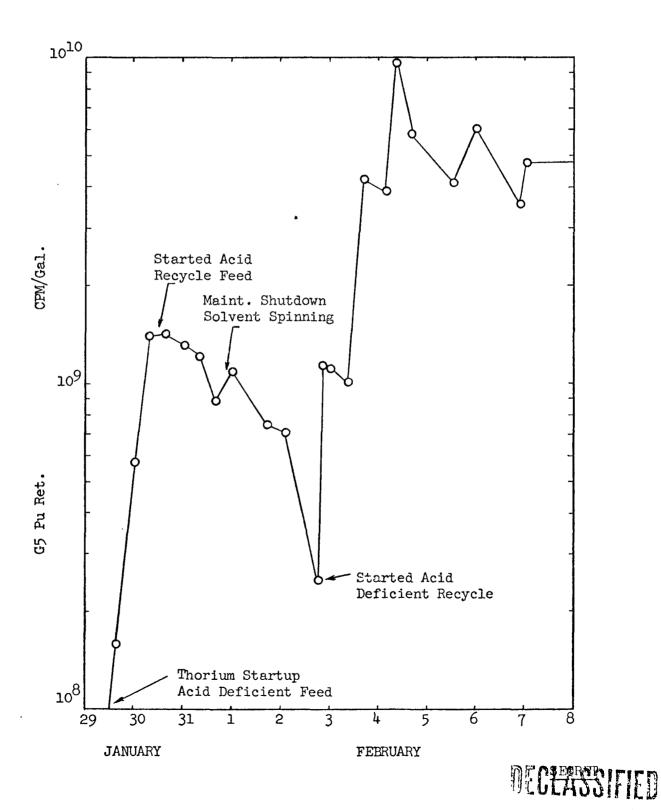
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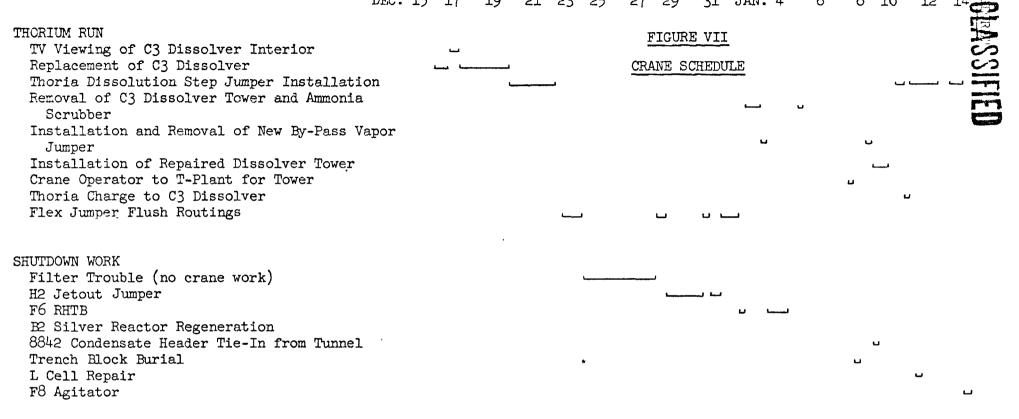
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FIGURE VI

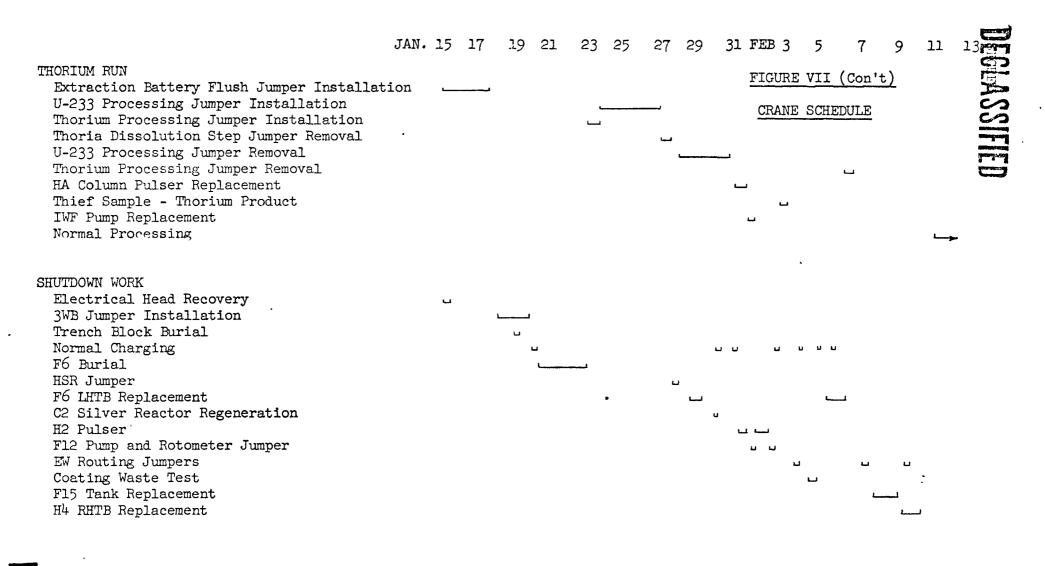




DEC. 15 17 19 21 23 25 27 29 31 JAN. 4 6 8 10 12 14



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TABLE I

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THORIUM MAJOR ROUTING & EQUIPMENT CHANGES

CANYON

Routes	Service	No. of Jumpers Required
C3-E3 E1-C3 E5-D2 C3-F12 E5-J1 3A Co1-F12 J2-7713 HDR 3B Co1-J5 R1A-J3 J3-J2 K1-J1	Dissolver to Centrifuge Feed Tank Centrifuge Slurry Tank to Dissolver Centrifuge Catch Tank to Coating Waste Tank Dissolver to IWF Tank Centrifuge Catch Tank to 3WB Tank 3AW to IWF Tank HAO Tank to No. 1 Organic HDR 3BN to 2AF Tank Organic Makeup Tank to 1BXF Tank 1BXF Tank to HAO Tank 2DF Tank to 3WF Tank	Required 2 3 2 4 3 1 1 1 1 2 2
Kl-Trailer Loadout HSIS-HNO ₃ -HS Col HAX-HNO ₃ -HA Col Kl-FlO IF Jetout Hdr-FlO HA Col-Fl2 Fl2-Fl6 E5-Fl2	2DF Tank to UNH HDR Utility Spare to HA Column HSR Nitric to HA Column 2DF Tank to 3WF Tank Interface Jetout Header to 3WF Tank HAW to IWF Tank IWF Tank to PAW Tank Centrifuge Catch Tank to IWF Tank	2 1 3 Flex 1 Flex 1 Flex 1 Flex 1 Flex

Equipment

New C3 dissolver installed.

Repaired dissolver tower installed.

Temporary tower drain blank installed.

Temporary C3 dissolver vapor by-pass jumper installed.

PIPE & OPERATING GALLERY

Routes	Service
3AF Acid-J5	3AF acid routed to 2AF tank at JG95
7007 HDR-K1	AMU utility header to 2DF tank at KG127

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TABLE I (Con't)

PIPE & OPERATING GALLERY (Con't)

Routes	Service
Chem Add Dolly-L3A	Chem add dolly fitting installed on 2BP stripper acid system.
Chem Add Dolly-C3	Chem add dolly fitting installed on dissolver nitrate add system.
Jet Steam-3BN Jet	3WB tank to 3WB concentrator jet steam supply used for 3BN jet.
7007 HDR-HS Col	Concentrated nitric header to HS Col at HG37
217TK-J5	2DF iron feed tank routed to 2AF tank at JG94
219 TK-HS Col	Auxiliary nitrite feed tank routed to $\mathrm{HSS-H_2O}$ line
Steam-C3 Sparger	Steam supply routed to C3 dissolver
UO ₃ HDR-U5 (Blanked)	UO ₃ recovered acid header to AFF tank blanked in pipe chase.

INSTRUMENT

3B and HA column Dp recalibrated.

293-A dissolver off-gas I_{131} monitor recalibrated for Ru.

C3 dissolver knock-out pot liquid level alarm installed.

C3 dissolver off-gas Ru monitor installed.

C3 dissolver coil discharge monitor and alarm installed.

C3 dissolver temperature probe and Sp.Gr. Wt.Ft. dip tubes shortened.

PSC concentrator (L4) high Sp.Gr. alarm installed.

3AF acid flow meter repositioned.

HSS-Butt rotometer installed.

HSIS-Nitric rotometer installed.

Pa load-out cask liquid level probe installed.

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TABLE II

DISSOLVER CLEAN-OUT SUMMARY

	01a C3 D	issolver	New	C3 Dissolve	er	
	Original	Modified	With Original	Tower	With Re-Used	
	Flushes	Flushes	Tower	Removed	Tower	Total
Number of Flushes	27	21	17	3	2	70
Time (Days)	29	18	14	4	2	69
Total Volume (Gal)	64,630	27,520	35,630	5,000	3,285	136,630
Chemical Cost	\$6,034	\$2,853	\$3,321	\$379	\$367	\$12,954
Uranium Removed (1bs)	79.8	7.8	8.2	0.14	0.87	96.8
Plutonium Removed (gm)	91.2	11.6	6.2	0.35	1.07	110.4
Pu/U (gm/ton)	2,286	2,974	1,512	5,000	2,459	2,280
Cs/U (curies/ton)		2.75x103	3.54x103	3.50x103	2.58x103	3.09x103
Avg. Uranium Dissolving Rate (gm/hr)	153.7	26.9	32.6	1.5	31.4	:
Avg. Corrosion Rate mils/mo.	8.5	14.9	4.6	l	5.6	

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TABLE III

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PRE-TEST DISSOLVER FLUSHING DETAILS

	TABLE III PRE-TEST DISSOLVER FLUSHING DETAILS											Corrosi and
Flush	Туре	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem	U Picked Up For Time Period (1b)	Pu	Pu/U gm/ton)	U Pickup Rate gm/hr	Corrosient) Rate mil/mo.
1	57% HNO3	2000	11-3	21	Boiling	B-3	335	26	24	1846	491	
2	57% HNO3	400	11-6	20	Boiling	F-12	67		6.4		4.5	
3	57% HNO3	400	11-7	20	Boiling	F-12	67	0.80	1.4	3500	18	
4	20% HNO3	2800	11 - 9	4	95°	F-12	136	1.40	1.2	1714	159	
5	10% NaOH-2% Tartaric	2800	11-10	4	800	D-2	254	4.7	3.3	1404	533	
6	30% HNO3-0.03M KF 0.15M ANN	2800	11-13	l	80 ⁰	F - 12	261	25	3 9	2720	11340	<25
7	Water	2800	11-15	l	80 °	F-12	•	0.90	1.3	2889	408	
8	20% HNO3	2800	11-16	4	80 ⁰	F-12	136	0.59	1.0	3390	67	
9	10% NaOH-2% Tartaric	2800	11-18	4	800	D-2	254	3.6	5.9	3278	408	
10	30% HNO3-0.03M KF 0.15M ANN	2800	11-19	l	80°	F-12	261	8.7	2.7	620	3946	~53
11	Water	2800	11 - 19	l	800	F-12		0.9	1.3	2890	408	
12	30% HNO3-0.03M KF 0.15M ANN	2800	11-20	4	Boiling	F - 12	261	1.8	1.2	1330	204	< 33
13	10% HNO3	2800	11-21	1	50 °	F-12	68	0.28	0.10	714	127	RL-SE Page 1
14 7100	30% HNO3-0.03M KF 0.15M ANN	2800	11-21	4	Boiling	F-12	261	0.95	0.45	947	108	RL-SEP-352 Page VI-15

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					TAE 3	🗧 (Con'	t)						ß
Flush	Туре	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem Cost (\$)	U Picked Up For Time Period (1b)	Pu	Pu/U (gm/ton)	U Pickup Rate gm/hr	Corrosion Rate 	Los Xour
15	Water	2800	11-22	l	800	F-12		0.21	<0.1	< 952	95		N.J
16	30% HNO3-0.03M NaF 0.15M ANN	2800	11-23	15	Boiling	F -1 2	261	0.89	0.48	1079	27	5	
17	10% NaOH-2% Tartaric	2800	11-24	4	800	D-2	254	1.02	0.11	216	116		
18	57% HNO3-0.025M NaF 0.125M ANN	1400	11-25	12	Boiling	F-12	257	1.57	0.34	433	59	4	
19	57% HNO3-0.025M NaF 0.125M ANN	1515	11 - 26	12	Boiling	F-12	257	0.34	0.35	2058	13	4.4	
20	57% HNO3-0.025M NaF 0.125M ANN	1580	11-28	8	Boiling	F-12	257	0.16	0.01	125	9	3.3	
21	57% HNO3-0.025M NaF 0.125M ANN	2192	11-28	8	Boiling	F-12	3 55	0.12	0.19	3167	7	3.1	
22	57% HNO3-0.025M NaF 0.125M ANN	1927	11-29	8	Boiling	F-12	323	0.15	0.04	533	9	4.3	
23	25% Caustic	1980	11 - 29	4	Boiling	D-2	205	0.55	0.21	764	63		
24	57% HNO3-0.025M NaF 0.125M ANN	2390	11-29	8	Boiling	F-12	376	0.11	0.14	2546	6	4.9	
25	57% HNO3-0.025M NaF 0.125M ANN	2185	11-30	8	Boiling	F-12	376	0.094			5	4.0	
26	57% HNO ₃ -0.025M NaF 0.125M ANN	2360	12-1	8	Boiling	F-12	376	0.09	0.08	5 1890	5	3.0	RL-SEP-352 Page VI-16
DECL	57% HNO3-0.025M NaF 0.125M ANN	2280	12-1	8	Boiling	F - 12	376	0.14	0.13	1860	8	5.7	P-352 VI-16
DECLASSIT	28% HN03-0.05M NaF 0.05M ANN	500	12-1	8	Boiling	F-12	37	0.14	0.01	4 2000	8	5.8	

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							TAB.	(Con'	t)					
Flush			Type	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem Cost (\$)	U Picked Up For Time Period (1b)	Pu	Pu/U (gm/ton)	U Pickup Rate gm/hr	Corrosion Rate mil/mo.
29	28%	HNO3.	-0.05M NaF 0.05M ANN	1000	12 - 2	24	Boiling	F - 12	74	0.82	1.10	2683	15	8.1
30 - A	. 20%	ниоз.	-0.025M NaF 0.025M Na2Cr207	450	12-5	8	450			0.147	0.14	1905	6	
30-E	3 20%	HNO3	-0.76M NaF 0.08M Na2Cr207	1400		4	60 0			0.453	0.21	927	51	
	20%	HNO3	-0.76M NaF 0.08M Na2Cr207			5	300	D-2	73	<0	<0		<0	1.8
31	Wate	er		1000	12-6	0	<50°	D-2						
32	15%	NaOH		500	12-6	l	Boiling	D-2	23					
33	12%	HNO3		320	12-6	l	Boiling	D-2	. 9					
34	57%	HNO3	-0.025M NaF	2160	12 - 7	0	350			0.07	0.03	7 1057		
			0.05M ANN	2153		4	Boiling			0.60	0.05	7 190	60	
				2166		~4	Boiling							:
	57%	HNO3.	-0.033M NaF 0.05M ANN	2205		4	Boiling			1.01	1.18	2340	51	4.8
				2170		4	Boiling			1.78	1.64	1840	57	8.8
				2135		4	Boiling			1.35	1.86	2760	25	30.0 PR
	57%	hno3.	-0.04M NaF 0.05M ANN	2150		4	Boiling			1.57	1.61	2050		RL-SEP-352 Page VI-17 0 0 0 4 0 2
				2170		4	Boiling			1.69	2.42	2860	32	52
DECLEASSIFIED	57%	HNO3	-0.049M NaF 0.05M ANN	2170		4	Boiling			1.82			15	48.0
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					TAE:	<u> </u> (Con'	t)					
Flush	Туре	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem Cost (\$)	U Picked Up For Time Period (1b)	Pu	Pu/U gm/ton)	U Pickup Rate gm/hr	Corrosion Rate <u>mil/mo.</u>
	57% HNO3-0.049 NaF 0.05M ANN	2170		4	Boiling			2.28			52	C,
	57% HN03-0.049 Naf 0.061M ANN	2213		6	Boiling			2.21	2.96	2680	<0	26.0
	57% HNO3-0.059M NaF 0.061M ANN	2221		4.	Boiling			2.04	3.33	3260	< 0	26.0
	57% HNO ₃ -0.059M NaF 0.067M ANN	2283		4	Boiling	D - 2	356	1.60	3.71	4640	< 0	31.2
35	Water	300	12-9	0	350	D-2						
36	1.2% HNO3	1543	12-10	l	Boiling	D-2	5	0.123	0.154	2570	56	
37	57% HNO3-0.025M NaF 0.05M ANN	2168		0	35°		•	0.028	0.035	2500		
		2168		4.	Boiling			0.607	0.473	1560	66	7.0
		2115		6	Boiling			0.466	0.782	3360		3.5
		2115		6	Boiling			0.634	0.910	2870	l	10.1
		2115		4	Boiling	D-2	348	1.04	1.270	2440	46	20.7
38 ⁽²	20% HN03-0.5M NaF	80	12-11	4	~800	D - 2	15	0.032	0.034	2125	3.6	9.8
₃₉ (2	2)20% HNO3-0.5M NaF	80	12-12	4	~80°	D-2	15	0.043	0.049	2279	5	18.0
0 ⁴⁰ ⁽²	2)20% HN03-0.5M NaF	80	12-12	42	2/3~800	D - 2	15	0.170	0.099	1176	16.3	31.0 ge
$\mathbf{D}^{\mu_1(2)}$	2)20% HN03-0.5M NaF	80	12 - 13	42	/3 ~80°	D-2	15	0.073	0.238	6520	7	RL-SEP-352 Page VI-18 31.9 .9
	2)20% HNO3-0.5M NaF 57% HNO3-0.025M NaF 0.05M ANN	2338	12-14	0	35 [°]			0.010	0.002	400		18 18
State - and												

					TABLE	III (Con't	;)						
Flush	n Type	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem	U Picked Up For Time Period (1b)		Pu/U m/ton)	U Pickup Rate gm/hr	Corrosi Rate mil/mo.	
	57% HNO3-0.025M NaF 0.05M ANN			412	Boiling			0.350	0.599	3420	35	< 3	
				6	Boiling			0.620	0.852	2750	47	< 0	
				6	Boiling	D-2	348	0.966	1.380	2860	27	<0	
	2)20% HNO3-1.0M NaF	80	12-15	4	80° → 35°	D-2	16	0.019	0.023	2868	2.1	61.5	
44(2	2)20% HN03-1.0M NaF	80	12-15	8늘	70°->95°->60°	D-2	16	0.131	0.312	4760	7	70.0	
. 45	57% HNO3-0.025M NaF 0.05M ANN	2260	12-16	0	350			0.017	0.073	8589			
		2260		4	Boiling			1.107	0.998	1800	125	17.0	
		2260		4	Boiling	D-2	* 348	1.197	1.081	1815	11	10.0	
46	57% HNO3-0.025M NaF 0.05M ANN	2237	12-16	0	350			0.057	0.073	2560			
		2183		24	Boiling	D-2	350	0.221	0.319	2890	41	<6	
47	57% HN03-0.025M NaF 0.10M ANN	2315	12-17	0	~35°			0.057	0.0732	2580			
				4	Boiling	D-2	350	0.221	0.319	289 0	20		
48(2	2)20% HN03-2.0M NaF	60	12-18	4	60° → 95°			0.0450	0.115	5111	5	<8.6	
DECLASSIFIED	53% HNO3-0.05M NaF 0.10M ANN	2360		7	Boiling	D-2	344	0.330	0.540	3270	31	RL-SEP-352 Page VI-19 © N	
Production of the second													

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TABL___I (Con't)

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1					TABL	<u>I</u> (Con'	c)					
					NEW DIS:	SOLVER INST	LLED					
Flus	n Type	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (^O C)	Disposed To	Chem Cost (\$)	U Picked Up For Time Period (gm)	Pu (gm)	Pu/U (gm/ton)	U Pickup Rate gm/hr	Corrosion Rate mil/mor
l	57% HNO3-0.025M NaF 0.05M ANN	2360	12 - 21	2 <u>1</u>	Boiling			281	0.753	2428	112.0	
		2390	•	3	Boiling			400	0.999	2763	39•7	16.1
		2390		20	50 0	D-2	348	541	1.51	2529	7.1	1.0
2	57% HNO ₃ -0.025M NaF 0.05M ANN	2362	12-22	0	350			8	0.023	6 2676		
		2002		2	Boiling			30	0.070	2116	11.0	
		1944		2	Boiling	D-2	348	123	0.263	1937	46.5	1.3
3	30% HNO3-Tower Flush	500	12 - 23	0	500		•	91				
		2000		1 3/4	Boiling			236	0.377	1450	82.9	
		22 00		l	500	D-2	168	100	0.192	1739	< 0	•
4	30% HNO3-Tower Flush	530	12 - 24	0	500			0.1				
		2053		3	Boiling	D-2	157	410	0.975	2155	137.0	
5	30% HNO3-Tower Flush	2129	12-24	2	Boiling	D- 2	163	58	0.144	2243	29.0	
6	30% HNO ₃ -Tower Flush	1767	12-24	3 <u>1</u>	Boiling	-		55.4	0.015	246	15.8	P R
)EC		1698		3	Boiling	D - 2	136	123.3	0.265	1947	22.6	E-SE
	30% HNO3-Tower Flush	1842	12 - 25	1 3/4	Boiling	D-2	147	82.8	0.190	2076	47.3	RL-SEP-352 Page VI-20
KIN 8	30% HNO3-Tower Flush	1861	12 - 25	2	Boiling	D-2	142	57.5	0.150	2363	28.8	U U
DECLASSITIED	30% HNO ₃ -Knockout Pot Flush	2108	12-26	5 3/4	Boiling	D-2	162	110.1 ,	0.290	2386	19.1	

TABLE III (Con't)

	Flush	Type	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (°C)	Disposed To	Chem Cost (\$)	U Picked Up For Time Period (gm)	Pu	Pu/U gm/ton)	U Pickup Rate gm/hr	Corrosion Rate mil/mo.
	10	30% HNO ₃ -Knock-out Pot Flush	1751	12-26	2	Boiling	D-2	135	95.4	0.168	1595	47.7	0
	11	30% HNO3-Knock-out Pot Flush	1951	12-26	3 <u>1</u>	Boiling	D-2	150	124	0.247	2004	35.4	
	12	57% HNO3-0.025M NaF 0.05M ANN	2385	12-26	0	50 ⁰			27	0.061	2049		
			2300		4	1000			87.5	0.239	2477	15.0	
			2300		1 2/3	Boiling	D-2	348	146	0.340	2105	34.4	
	13	30% HNO3-Knock-out Pot Flush	2260	12 - 27	4 <u>1</u>	Boiling	D-2	174	84	0.020	205	18.7	
	14	57% HN03-0.025M NaF 0.05M ANN	2345	12 - 27	0	50 ⁰		*	11.7	0.025	194		
			2345		2	Boiling			127.5	0.213	1515	58.0	
			2345		6 3/4	Boiling	D-2	348	148.7	0.427	2604	2.9	
	15	57% HNO ₃ -0.025M NaF 0.05M ANN	2192	12-29	81	90 ⁰			59.0			6.9	3.1
			2192		6	90 0			218.9	0.291	907	15.1	7.3
	J 7		2115		5	Boiling	D-2	348	364.1	0.792	1973	29.0	7.9 Page
U	2 16	Water + 1500 gal. 57% HNO ₃ through Tower	2315		2				305	0.689	2049	152.5	age ∧ SET
S		mog om oden tower			6	Boiling	D-2	27	406	1.162	2861	16.8	-SEP-352 3e VI-21
JLAN SIFILU	17	Water + 850 gal. 57% HNO ₃ through Tower	1750		2	Boiling			159	0.512	3223	78.5	

BEOLASS

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					TABL	[(Con't	t)					Notes they are the second seco
Flush	Туре	Vol. (Gal.)	Date	Time (Hrs)	Temp. For Time Period (OC)	Disposed To	Chem Cost (\$)	U Picker Up For Time Period (gm)	Pu	Pu/U gm/ton)	U Pickup Rate gm/hr	Corrosion Rate mil/mo.
	5 Cycles-Boiling-Cool	1628		~3	Boiling			339	0.797	2352	60.0	Const.
	5 Cycles-Boiling-Cool	1628		~3	Boiling			458	1.137	2482	39•7	
	5 Cycles-Boiling-Cool	1600		~3	Boiling	D-2	20	679	1.385	2040	73.7	
₁₈ (3)5% HNO3	1410	1-3	0	35°			14.7	0.052	3568	•	
		1065		3	95 ⁰	D-2	15	19.3	0.065	3353	1.5	
19	5% HNO3	1500	1-4	0	350			1.8	<0.002	<987		
		1200		3	95°	D-2	16	3.6	0.012	3000	0.6	
20	57% HNO3-0.025M NaF 0.05M ANN	2100	1-5	28	Boiling	D-2	348 •	43.7	0.275	5708	1.6	<1.7
21(4)57% HN03-0.025M NaF 0.05M ANN	2178	1-9	0	35°			50				
		2178		2	Boiling			267	0.756	2832	108.5	
		2150		5	Boiling	D-2	348	322	0.813	2525	31.0	5.6
22	57% HNO3 through Tower	1676	1-10	2	Boiling			47	0.•069	1459	23.5	
		1360		2	Boiling	D-2	19	74	0.266	3600	13.5	

 (1)Corrosion rate based on surface area of coil exposed to boiling conditions. Where coil is not affected such as flushes 43 and 44 rate is based on surface area exposed to the solution.
 (2)ANN added on a 2-1 mole ratio to NaF after heating period, also 500 gallons of water was added to get a flushes 43 and 44 rate is flushes 43 and 44 rate is (2)ANN added on a 2-1 mole is volume to sample. (3)Dissolver tower removed. (4)Re-used tower installed.

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TABLE IV

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	SUMMARY OF	SOLVENT EX	TRACTION F	LUSHES
Equipment Systems Flushed	Number of Flushes	Total Vol (Gal)	Chem Cost (\$)	Type(2) of Flush
L Cell Package	5	400	31	30% HNO ₃
J5-L1-I2	5 4 2	25,000 2,000 1,000 1,000	46 91 151	Water 10% HNO ₃ Caustic ³ Tartaric Oxalic Nitric
J Cell Package	5 5 2 2	3,500 3,500 1,400 1,400	78 129 208	Water 10% HNO ₃ Caustic Tartaric Oxalic Nitric
J1-н2-н3- J3-J6-J7-J8-К1	5 4 6	25,000 20,000 30,000 30,000	666 3,750 5,698	Water 20% HNO ₃ Caustic Tartaric Oxalic Nitric
KI-JI KI-F10	12	49,000		Water
J7-J8-K1-J1	2	13,200		Water
J 6-J4-J2-J5	3	9,600	220	10% HNO3
G&R Cell	2 2 7	1,000 9,000 31,500	540 5,746	Water Caustic ™artaric Oxalic Nitric
TOTAL(1)	81	257,100	17,354	

(1)Total includes 31 water flushes with a total volume of 116,700 gallons. (2)Concentration 10% Caustic-2% Tartaric and 5% Oxalic-30% Nitric

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TABLE V

THORIUM DISSOLVING AND DENITRATION

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<u>Cut No.</u>	Dissolving Time Hrs	Denitration Time Hrs	Thorium Dissolved #	Thorium Dissolving Avg. Rate (lb/hr)	Acid(1) Used Gals	Cost Per Batch \$
l	33	₂₂ (2)	3000	91	4500	770
2	60	23	2500	42	6000	1020
3	46		2050	44	5000	850
4	34		1080	31	4500	770
5	36	₃₆ (3)	470	13	5000	850
Sub-Total	209	81	9100		25000	4260
6	23		170 ·	7.4	2000	350
7	8		50	6.3	2000	350
8	9		17	1.9	2100	350
9	20		8	0.4	1500	250
10	55		8	0.15	2300	390
Total	324	81	9353		34900	5950

(1)Initial volume 2000 gallons, 57% HNO3, 0.025M NaF, 0.05M ANN for cuts 1-7. Additional acid added in 500 gallon increments. Flushes 8-10 57% HNO3.

(2)Completed first cut denitration with second cut.

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(3)Third, fourth and fifth cuts combined for denitration.





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TABLE VI

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PROTACTINIUM RECOVERY

		Time (Hrs)	Vol. (Gal)	Chem Cost _(\$)	Pa Recovered (gm)
Pa Run on U ²³³	lst Pa Recovery	26	2,000	215	17
Feed Solution (~21 gms)	Ba SO4 PPT. Removal	13		0	~5(1)
	Final Pa Recovery	10		35	4 <u>+</u> 2
	Sub-Total	49		250	4 <u>+</u> 2
Pa Run on Th Waste (20 gms ± 5 gms)	Two Attempted Pa Recovery Runs	48 •	6,000	655	< 10
Total Pa Recovery		97		905	4 ± 2

(1)A portion of Pa lost to solvent extraction feed due to equipment failure.

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TABLE VII

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U233 MATERIAL BALANCE

Dissolver Input

3276 gm

U^{233} Run Feed	3090 gm		
Loadout			2886 gm
Batch 1 Batch 2 Batch 3 Batch 4 Batch 5 Flush 6 Flush 7 Flush 7 Flush 8 Flush 9 Flush 10 Flush 11	324 gm 428 gm 852 gm 689 gm 413 gm 59 gm 31 gm 46 gm 18 gm 21 gm 6 gm	·	

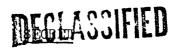
390 gm

Losses

Coats	66 gm	
Dissolver Heel (6th & 7th cuts)	70 gm(1)
Waste (Total)	126 gm	
Sumps (Total)	128 gm(2)

(1)Lab Analysis 115 gm
(2)Lab Analysis 377 gm

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TABLE VIII

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THORIUM MATERIAL BALANCE

Input		9489 lbs
Th Run Feed 9100 1	bs	
Loadout		7660 lbs
Losses		1829 lbs
Coats	200 lbs	
Dissolver Heel (6th & 7th Cuts)	220 lbs	
Waste (Total)	975 lbs ⁽¹⁾	
Sumps	434 lbs ⁽²⁾	

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(1)Lab Analysis 802 lbs (2)Lab Analysis 350 lbs

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TABLE IX

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SOLVENT EXTRACTION THORIUM REMOVAL FLUSHES

Equipment Systems Flushed	No. of Flushes	Туре	Total Vol (Gal)	Chem Cost (\$)
н2-н3	3	Oxalic Nitric (1)	9,000	1,360
J 3- J6-J4-J5	2	10% HNO3	8,000	184
J7-J8-K1	4	Water	7,000	
J21- J22-J23	5	10% HNO3	2,500	58
J5-L1-L2	5	10% HNO3.	6,000	138
L3-L4-L6	5	10% HNO3	150	4
G Cell	7	Oxalic Nitric ⁽¹⁾	18,500	2,794
TOTAL	31		51,150	4,538

(1)Concentration 5% Oxalic-30% Nitric.

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TABLE X

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	Tł	IORIUM RUN WA	STE ROUTES A	IND VOLUMES	_)	
Originating Tanks	Nov Vol (Gal)	Dec Vol (Gal)	Jan Vol (Gal)	Feb Vol (Gal)	Total	Receiving UGS
D2	11,000	119,700	24,100	22,200	177,000	1020
F16-F18-E5	15,000	98,800	409,100	42,300	565 , 200	105A
G8-R8		62,100	58,300	210,000	330,400 (101-103-104-106 A
TOTAL	26,000	280,600	491,500	274,500	1 ,0 72,600	

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(1)These volumes represent neutralized waste transferred to UGS originating from plant flushing and thorium processing. They do not include wastes generated from normal processing in Nov., Dec., and Feb.

SKETCH



THORIUM PROCLSS FLOW SKETCH

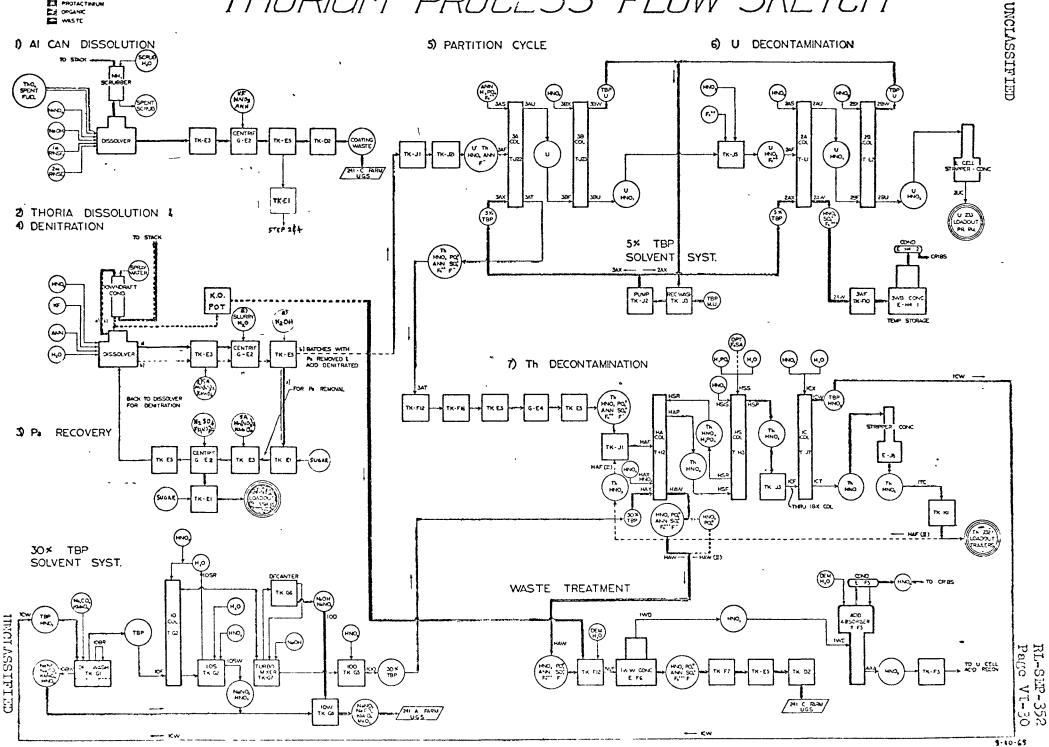
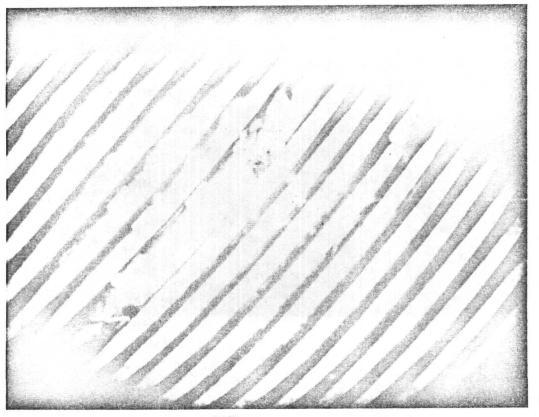


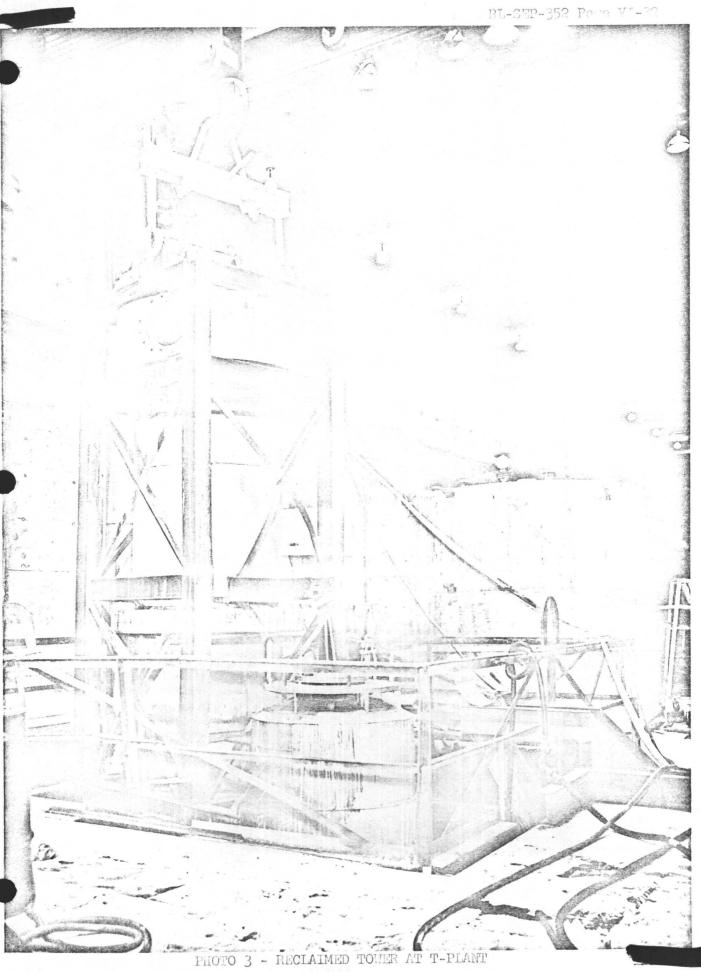
PHOTO 1 - TV VIEWING OF C3 DISSOLVER

PHOTO 2 - SCRAP ON DISSOLVER GRID



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