DOE/ET/34022-1 MITNE-234 MIT-EL-80-032

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ANALYSIS OF STRATEGIES FOR IMPROVING URANIUM UTILIZATION IN PRESSURIZED WATER REACTORS by

> Joseph A. Sefcik Michael J. Driscoll David D. Lanning

DOE Contract No. DE-AC02-79ET34022 (Nuclear Engineering Department Report No. MITNE-234, Energy Laboratory Report MIT-EL-80-032). January, 1981.

DOE/ET/34012-1

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Joseph A. Sefcik Michael J. Driscoll David D. Lanning

Annual Technical Progress Report/Principal Topical Report For FY 1980

> Department of Nuclear Engineering and Energy Laboratory MASSACHUSETTS INSTITUTE OF TECHNOLOGY Cambridge, Massachusetts 02139

> > Sponsored by

U.S. Department of Energy Division of Energy Technology Under Contract No. DE-AC02-79ET34022

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Printed in the United States of America

Available from

National Technical Information Service U.S. Department of Commerce 5285 Port Royal Road Springfield, VA 22151 USDOE LWR TECHNOLUGY PROGRAM IMPROVED URANIUM UTILIZATIUM SPECIALIZEU DISTRIBUTION LIST

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ABSTRACT

Systematic procedures have been devised and applied to evaluate core design and fuel management strategies for improving uranium utilization in Pressurized Water Reactors operated on a once-through fuel cycle. A principal objective has been the evaluation of suggested improvements on a self-consistent basis, allowing for concurrent changes in dependent variables such as core leakage and batch power histories, which might otherwise obscure the sometimes subtle effects of interest. Two levels of evaluation have been devised: a simple but accurate analytic model based on the observed linear variations in assembly reactivity as a function of burnup; and a numerical approach, embodied in a computer program, which relaxes this assumption and combines it with empirical prescriptions for assembly (or batch) power as a function of reactivity, and core leakage as a function of peripheral assembly power. State-of-the-art physics methods, such as PDQ-7, were used to verify and supplement these techniques.

These methods have been applied to evaluate several suggested improvements: (1) axial blankets of low-enriched or depleted uranium, and of beryllium metal, (2) radial natural uranium blankets, (3) lowleakage radial fuel management, (4) high burnup fuels, (5) optimized H/U atom ratio, (6) annular fuel, and (7) mechanical spectral shift (i.e. variable fuel-to-moderator ratio) concepts such as those involving pin pulling and bundle reconstitution.

The potential savings in uranium requirements compared to current practice were found to be as follows: (1) 0-3%, (2) negative, (3) 2-3%; possibly 5%, (4) 05%, (5) 0-2.5%, (6) no inherent advantage, (7) 00%. Total savings should not be assumed to be additive; and thermal/hydraulic or mechanical design restrictions may preclude full realization of some of the potential improvements.

ACKNOWLEDGEMENTS

The work presented in this report has been performed primarily by the principal author, Joseph A. Sefcik, who has submitted substantially the same report in partial fulfillment of the requirements for the Ph.D degree in Nuclea Engineering at M.I.T.

The present work was supported by a Department of Energy (DOE) contract administered through the M.I.T. Energy Laboratory. Computer calculations were carried out at the M.I.T. Information Processing Center and the Laboratory for Nuclear Science.

The authors would like to acknowledge the cooperation of Dr. Edward Pilat and employees of the Yankee Atomic Electric Company in facilitating the acquisition of descriptive data from publicly available documentation on the Maine Yankee Reactor. The information in this **re**port cited as deriving from "Maine Yankee", however, should not be considered as representing that actual system in its current or projected operating configuration, but as an idealization thereof. In particular, the results so identified in this report have not been either reviewed or approved by the Yankee organization.

The majority of the typing was ably handled by Ms. Maureen Walsh, with the assistence of Ms. Carolyn Rose and Ms. Christal Whelan.

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CHAPTER 1

BACKGROUND AND PREVIOUS RESEARCH

1.1 INTRODUCTION

The efficient use of fuel resources is an important goal for any energy technology. This objective has been assigned higher priority in the past several years with respect to natural uranium consumption by light water reactors (LWR's). Several reasons have motivated this attention: the rising price of uranium on the domestic and world market; the deceleration of the breeder development/ deployment program; and the deferral of plans to recycle plutonium in LWR's. The deferral of recycle is largely due to concerns over linkage to weapons proliferation.

The work described in this report was undertaken under the LWR Technology Program for Improved Uranium Utilization sponsored by the United States Department of Energy. It continues earlier MIT efforts along the same general lines undertaken as part of the Nonproliferation Alternative Systems Assessment Program (NASAP) (N-1) which in turn, provided input to the International Nuclear Fuel Cycle Evaluation (INFCE) (I-1). The general goal of the present work has been to develop, test, and apply selfconsistent methods for the evaluation of improvements in core design and fuel management strategy in pressurized water reactors as they apply to increasing the energy generated per unit mass of natural uranium mined.

The MIT effort has focused on pressurized water reactors (PWRs) because they comprise nearly 2/3 of the nuclear power plant capacity operable, under construction, or on order in the U.S. and over 1/2 of the worldwide nuclear capacity (see Table 1.1). However, it is expected that much of the work can be readily extended to the boiling water reactors (BWRs), which account for most of the remaining capacity. Similarly, the emphasis has been on concepts which can be retro-fit into current system designs, operating on the once-thru fuel cycle, since only items in this category have credible prospects for making a substantial impact through rapid and widespread deployment.

1.2 BACKGROUND

Several major studies have been completed over the past few years which delineate the potential uranium savings which can be achieved in pressurized water reactors. Table 1.2 summarizes some of the uranium saving techniques which have been identified, the potential savings, and the time interval needed for their introduction. The uranium-conserving strategies range from very simple to very complicated, and even though large savings are available in some cases (e.g. through reprocessing), the economics and politics of the situation may dictate against their use. It should also be noted that the various sav-

TABLE 1.1 NUCLEAR POWER PLANTS

(Operable, under construction, or on order (\geq 30 MWe) as of 6/30/80)

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TYPE (COOL/MOD.)		.S.			ORLD
PWR		113	(63.5%)		280	(53.2%)
BWR / LWR (H ₂ C)	61	(34.3%)		119	(22.6%)
LWBR						
PHWR (CANDU)					36	
LWCHWR					2	
HWBLWR	(D ₂ 0)				2	(8.0%)
					2	
GCHWR						
GCR				1	36	ļ
AGR					15	
LGR GRAPHITE		1			23	(14.4%)
HTGR	•	1			1	
THTR					1	
LMFBR (Na)		1			8	
		1.80				
TOTAL UNITS		178			526	
TOTAL GWE		171			400	
TOTAL OPERABLE		74			229	
GWE OPERABLE		54			125	
BWR = PHWR = LWBR =	Boiling Pressur Reactor Light W Light W	Wate: ized ater ater	- Breeder Re	er Moderat		and Cooled lerated
	Reactor Heavy W Cooled	ater I		Boiling I	⊿ight	: Water
				r Moderate	ed Re	eactor
GCR =	Gas Coo	led Re	eactor			
			-Cooled Re			
						ated Reacto
				Cooled Rea		-
				ire Reacto		_
			Cooled Fa	ast Breede	er Re	eactor
Source: Refere	nce (N-	2)				

	<u>PWR</u>	LIST OF SOM	E STRATEGIES	FOR FUEL CONSERVATION
Tactic	Percent Saving (approx.)	Time to Introduce	References	Comments
Reprocessing	20-30	10 years	G-2	Economic uncertainties Politically undesirable under current conditions
Increased Number of Batches	10	5 years	F-1,D-1, S-2	Constant discharge burnup would require more frequent refueling and would decrease capacity factor
Increased Discharge Burnup	10	5-15 years	F-1,D-1, S-2,M-2	Must assess materials problems
Reduced Radial Leakage	5-7	3-5 years	D-1,S-2, S-3	Commonly known as low-leakage fuel management; adverse power peaking is created
Axial Blankets	1-3	3-5 years	K-1,D-1, S-2	Leads to power peaking problems
Reduced Axial Leakage	1-2	5-10 years	K-1	Requires use of zirconium core support structures
Routine Coastdown	5-15	l-2 years	F-1,D-1, M-1,S-2, M-2,D-2 L-1	High savings if coastdown is done to economic breakeven
Variable Fuel to Moderator Ratio	5-15	l0 years	R-1	Ore savings depend on strategy used (e.g. pin pulling); may lead to thermal-hydraulic prob- lems. fuel reassembly required

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TABLE 1.2 REPRESENTATIVE LIST OF SOME STRATEGIES FOR FUEL CONSERVATION

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TABLE 1.2 (Cont.)

Tactic	Percent Saving (approx.)	Time to Introduce	References	Comments
D ₂ O Spectral Shift	10-15	10 years	C-1,G-1	Poison control is reduced; D ₂ O is expensive and losses are significant
Optimized Fixed Fuel/Moderator ratio (e.g. change lattice pitch, pin diameter or use annular fuel)	~2-3	5 years	R-1,B-1	Applicability depends on spe- cific reactor design

.

ings may not be additive when combined into a single design.

Current pressurized water reactors operate on a cycle which is one year in length, following which they are shut down for about six weeks for refueling. The mean discharge burnup is about 33000 MWD/MT. Every year one third of the core is replaced with fresh fuel which has an enrichment of about 3 w/o U-235. The fuel design characteristics and operating conditions of a reference PWR (Maine Yankee) are listed in Appendix A.

Increasing the number of staggered in-core batches results in a decrease in the cycle reactivity swing, which decreases the uranium enrichment required to meet the same energy output. The increased number of batches necessitates an increase in the number of refuelings in a given calendar period. An increased refueling frequency decreases plant capacity factor and thereby adversely affects the cost of electricity.

If the number of batches were held constant and the reload enrichment raised, the discharge burnup would be increased and the time between refuelings would be increased. The higher reactivity of the fresher batches permits the older batches to be driven deeper along the reactivity vs. burnup curve. Current pressurized water reactor owners are seriously considering switching to 18 month cycles with discharge burnups in the vicinity

of 36000 MWD/MT. The attendant increase in capacity factor helps reduce the cost of the generated electricity.

Employing both higher discharge burnup and an increased number of batches (e.g. 60000 MWD/MT, five batches) would reduce yellowcake consumption by 20%. There are potential difficulties, however, in implementing extended burnup, which are primarily related to the structural components of the fuel. The cladding faces increased corrosion on the water side during the long residence times and an increased internal pressure from fission product gas buildup. In-reactor experiments are under way at this time to assess the consequences of driving fuel to higher burnup.

Axial and radial leakage can account for neutron losses as high as four or five percent of total neutron production. Various schemes have been proposed for minimizing the losses. "Low-leakage" fuel management is a strategy whereby the fresh fuel is not loaded on the core periphery (as has been customary), but is instead scattered among older fuel in the core interior, while once or twice burned fuel is loaded on the periphery. The reduced peripheral power results in lower neutron losses, but higher radial power peaking factors throughout the reactor's burnup cycle.

The use of axial blankets of natural uranium or enrichment zoning in the axial direction can reduce axial leakage but, once again, power peaking factors are in-

creased. Neutron losses can be reduced without adversely affecting radial and axial peaking factors if stainless steel core support structures and/or core shrouds and barrels are replaced with zirconium structures.

Coastdown is a technique which relies on a reduction of moderator and fuel temperatures of the reactor, by reducing the coolant temperature and/or reactor power level to insert positive reactivity and thereby extend the length of the operating cycle. The electrical power output of the reactor is decreased in a ramp or stepwise fashion until the coastdown is terminated. The length of the coastdown period can be optimized to maximize the overall economics of the utility's electrical grid. Coastdown always reduces $U_3 O_8$ consumption. Under economically optimum conditions the reduction typically ranges from 5 to 8 percent. Coastdown to economic breakeven generates larger savings, around 12 percent.

The D_2O spectral shift concept uses a mixture of light and heavy water to change the neutron spectrum during the reactor's operating cycle, substantially reducing boron control requirements. At the beginning of the cycle, a 50% H_2O , 50% D_2O mixture is introduced into the reactor and the harder (more epithermal) spectrum increases absorption in U-238 which decreases reactivity in the short term but increases Pu-239 production. H_2O is added during the cycle to provide the reactivity needed to compensate for burnup. In-

decreased plutonium production and decreased losses to boron control poisons account for the improved neutron economy and increased fuel efficiency. D_2O is expensive, however, and losses can be quite costly. The separation of the H_2O and the D_2O may be more difficult than a utility would desire. Increased tritium production would complicate plant operation and maintenance. Hence, overall economics may preclude the use of this scheme.

Variable fuel-to-moderator ratio schemes seek to accomplish the same spectral shift as the D_2O-H_2O method but by mechanical means. The fuel begins in a tight pitch configuration with a lower reactivity than normal and an increased conversion ratio. At end-of-cycle, the rods are moved in some fashion to create a "wetter" configuration and in this manner, reactivity is inserted. Several of these schemes call for mechanical movement of the fuel, or fuel bundle disassembly and reassembly. The uranium savings are dependent upon whether the mechanical manipulation is done before or after the refueling (as will be discussed in a later chapter of this report). The thermal-hydraulic problems of both the fresh and reconstituted lattices involved in these schemes for producing a mechanical spectral shift may be severe.

The lattices of current generation pressurized water reactors were optimized under different constraints than those which apply at present: ore was cheaper and recycle

was assumed. As a result, the lattices may be slightly "dry" relative to the optimum fuel to moderator ratio for a high-burnup once-through cycle. A slight wettening of the lattice (through the use of annular fuel, reduced fuel density, etc.) may permit better fuel utilization on the once-thru cycle without violating safety criteria such as the need for a negative moderator temperature coefficient.

1.3 RESEARCH OBJECTIVES

To progress beyond the obvious gains obtained from increasing burnup and the number of fuel batches, it is generally believed that a large number of small, incremental improvements must be accumulated. This puts a severe strain on computational methodology: simple methods are too crude to discern the often subtle effects at issue and the use of full-fledged state-of-the-art computer capabilities involves far too much detail to be economically justified and, more importantly, admits so much increased flexibility that one can never be sure that comparisons are being made on an "all-else-being-equal" basis. Apparently innocuous simplifications or constraints can often perturb the comparison to the point where the effect of interest can be obscured.

This fundamental dilemma became clear early in the course of the present work, and led to the identification of the first major subtask -- the development of a simple scoping model which was capable of evaluating small changes

in core design and fuel management on a consistent basis and in a transparent manner. As developed, the model relies heavily on an analytically improved version of the linear reactivity model of core behavior, and on heuristic models benchmarked against more detailed computer results.

The second major focus of the present work has been on the evaluation of a number of contemporary candidate changes which have been suggested as ways to improve uranium utilization: the use of annular fuel pellets, axial and radial blankets, low leakage fuel management, and variable fuel-to-moderator-ratio concepts in several guises, such as pin-pulling and bundle reconstitution.

1.4 ORGANIZATION OF REPORT

The work reported here is organized as follows. Chapter two provides an outline of the computer methods used in this research. The calculational models used in these programs are also be discussed along with their applicability and limitations.

Chapter three contains an introduction to the linear reactivity model of core behavior. The model is extended to include the effects of leakage and power sharing as they occur in actual LWR fuel management schemes. An optimization process is developed and used to demonstrate that spatial and temporal criteria exist which can improve uranium ore use.

In chapter four, the use of simplified "group-andone-half" methods and some heuristic observations of computer-calculated reactor core maps leads to the development of simplified power sharing and core leakage algorithms for inclusion in the methodology developed in the previous chapter.

Chapter five reports on the application of the subject methods to axial fuel management. The basic analytic relationships and static bundle results are examined to determine the best approach for more sophisticated computer calculations. Several of the most promising candidates are examined in detail.

In chapter six the methodology is applied to discern the relative advantages of low-leakage fuel management and natural uranium radial blankets. Several core configurations are examined.

Chapter seven deals with variable fuel-to-moderator ratio concepts. The analytical additions and extensions of the advanced linear reactivity model necessary to calculate uranium savings for a wide range of options are developed. In addition, some modifications to the LEO-PARD computer program are required to adequately model the spectral shift concepts. Optimization criteria are presented and a final design proposed, one that is consistent with current perceptions of the thermal-hydraulic limitations.

Chapter eight summarizes the present work: the methods development effort and the several applications. The potential for the application of the same approach to other uranium saving strategies not analyzed here is discussed.

Several appendices are included to provide detail and supporting evidence and to document the computer program used to implement the acreening of the various fuel design options.

CHAPTER 2

COMPUTER METHODS

2.1 INTRODUCTION

A central problem of nuclear reactor fuel management i the accurate determination of reaction rates and isotopic d tributions at all points in space and time. The developmer of large and complex computer programs has simplified this task greatly. It is possible, in principle, to calculate the flux and isotopic concentrations in a nuclear reactor by using hundreds of energy groups, thousands of spatial points, and a multitude of time increments. However, practical considerations such as the size of today's computer memories and the costs associated with running long programs intervene to impose various constraints.

The evolution of computer programs in the nuclear fuel management sector reflects a continuing reduction of conservatisms imposed by safety requirements by a trade-off betwee cost and accuracy. Fortunately, acceptably accurate answers can be obtained without the necessity of complete pointwise modeling of the core. Thus, available "state-of-the-art" programs could be used to provide a sound foundation for the recent work.

The computer codes used in this research (primarily LEOPARD and PDQ-7) are employed either individually or in combination to make comparisons between various fuel manage-

ment strategies. The codes have been benchmarked against experimental data, and for certain regions of interest can be considered quite accurate.

Since there are several references which describe the methods associated with LWR fuel management (A-1,D-3,H-1,D-3) we can be brief in regard to generalities and concentrate on specifics in the exposition which follows. Even more germane, previous workers at MIT: namely, Garel (G-2), Fujita (F-1), and Correa (C-2) have used these same codes and documented both the general features of the programs and their experience with their use for applications similar to those of present concern.

The methodology of LWR fuel management can be thought of as two coupled, yet distinct, phases: the neutron spectrum calculation and the neutron diffusion calculation as depicted in Figure 2.1. In the pressurized water reactor, the broad features of the ambient neutron spectrum (e.g. fast, epithermal, thermal regimes) within the confines of a given fuel assembly depend primarily on burnup status. Thus, for a given fuel design at a specific burnup, the few-group crosssections and the concentrations of the isotopes, along with their attendant neutron spectrum, may be considered to be "state or state vector functions" of the burnup.

In non-fuel regions the cross-sections will not vary appreciably over the life of the fuel. In common practice, the cross-sections for both types of regions are generally "collapsed" into only two groups (fast and thermal) from the

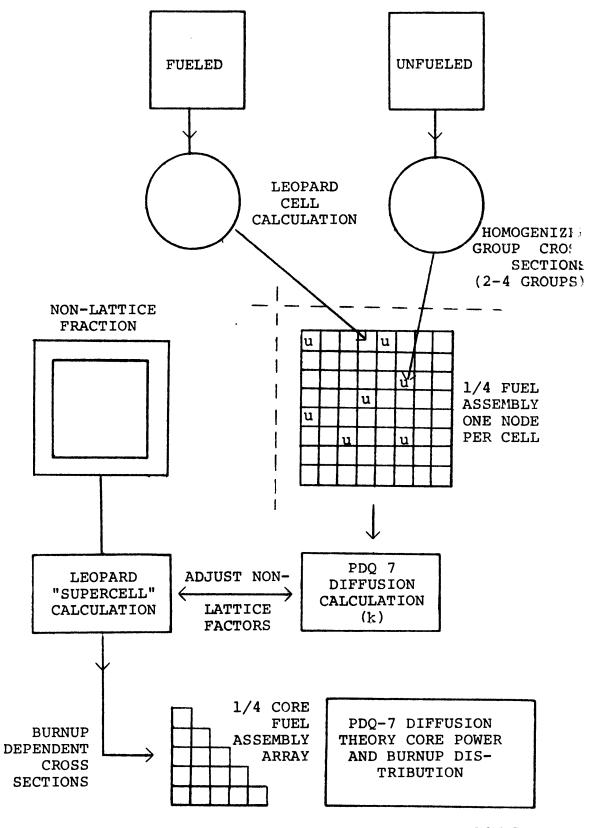


FIG. 2.1 FLOW CHART FOR LEOPARD AND PDQ CORE POWER CALCULATIONS

several hundred energy groups used in the unit cell spectrum calculation.

The burnup achieved at each point in the reactor depends on the local neutron flux level, and hence on local and near-neighborhood power levels. The flux level is determined by solving the spatial finite-difference equations (using the two-group spectrum) over a given time period. At each burnup time step the fine group spectrum calculation of the unit fuel cell is used to provide the collapsed two-group cross-sections used by the spatial calculation and the spatial flux shape is solved for again. The new flux levels are then applied once again to deplete the isotopes appropriately at each point. The process is continued in a successive fashion until the target energy production is achieved.

2.2 The LEOPARD Code (B-2)

The LEOPARD code calculates the spectrum for a "unit fuel cell" consisting of fuel, gap, cladding, moderator, and, in some cases, an "extra" region representing fuel assembly structure. In addition, neutron cross-sections can be calculated for unfueled regions. The cross-sections for the EPRI-LEOPARD version used in the present work are obtained from ENDF/B-IV data sets.

Material compositions, number densities, geometry, temperatures and pressures are input parameters to the code. A "zero-dimensional" (homogenized) calculation is performed to

determine the spectrum. An input buckling can be included to account for global leakage effects. In the thermal regime ($E_n < 0.625 \text{ eV}$) 172 energy groups are calculated over a Wigner-Wilkins spectrum using a SOFOCATE-type treatment (A-2), and disadvantage factors are calculated by a modification of the ABH method (A-3).

The fast spectrum calculation (in 54 groups) uses the self-consistent B-l approximation in the MUFT treatment (B-The resonance parameters are calculated using the procedures described by Strawbridge and Barry (S-4).

The few group cross-sections (2-4 groups) generated by LEOPARD apply to a properly "homogenized" cell and hence can be input directly into coarse spatial mesh calculations.

The cross-section sets used in this research are either fast and thermal, or fast and thermal mixed number density (MND) (B-4). The MND model uses neutron activation continuity as a boundary condition rather than flux continuity. The thermal ($E_n < 0.625 \text{ eV}$) values of Σ_a and Σ_f are averaged over a Wigner-Wilkins spectrum, whereas the thermal value of the diffusion coefficient, D, is averaged over Maxwellian spectrum.

The LEOPARD unit cell geometry is shown in Fig. 2.2. At the beginning of each time step, the neutron spectrum is recomputed and isotopic depletion during a time step is based on this spectrum. Boron concentration can be varied as a function of time, as can the D_2O/H_2O ratio to model spectral

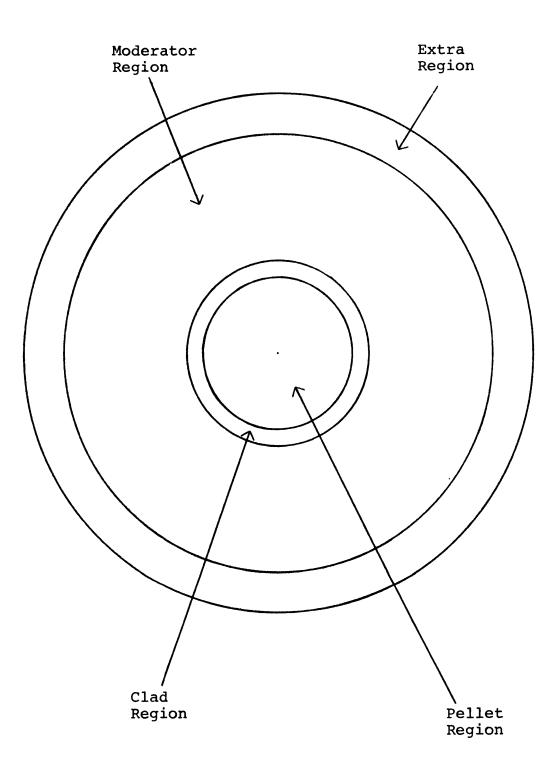


FIG. 2.2 LEOPARD UNIT CELL GEOMETRY

shift effects. A fission product scale factor can be adjusted to allow for the effects of variation in uranium or plutonium isotopic content. A discussion of some of the characteristics of the LEOPARD results (and some salient operational points) is included in Appendix B.

Correa (C-2) has benchmarked EPRI-LEOPARD (with its ENDF/B-IV based cross sections) at MIT. Whereas the primar emphasis of the calculations was tight pitch U-233/ThO₂ and Pu/ThO₂ cores, twenty-six U-235/UO₂ lattices were examined. The average k value for the critical experiments and the range of parameters investigated is shown in Table 2.1.

2.3 The CHIMP Code

The large number of flux-weighted microscopic crosssections produced by LEOPARD at each time step are processed by the CHIMP code, which was developed by the Yankee Atomic Electric Company (C-2). The LEOPARD output at each time step can be routed directly to cards, tape, or disk and this data can be directly manipulated by the CHIMP code to prepare cross section sets usable by PDQ-7, the spatial depletion code used in this research and described later in this chapter.

2.4 The HAMMER System (S-5)

The HAMMER system is a set of linked reactor physics programs for the calculation of infinite lattice parameters using multigroup transport theory, and composite reactor

TABLE 2.1 SUMMARY OF BENCHMARK COMPARIS

Fuel:	U-233/ThO ₂	U-235/ThO ₂	U-235/UO ₂	U-235/U	U-235/UO ₂	Pu/UO ₂
ε (w/o)	3.00	3.78 - 6.33	3.00 - 4.02	0.7 - 1.5	1.3-4.1	1.5-6.6
F/M	0.01-1.00	0.11 - 0.78	0.23 - 2.32	0.15 - 1.69	0.1-1.3	0.1-0.9
(H+D)/U-238 (or/Th-232)	3.4 - 403.	4.7-36.	1.31 - 14.6	0.8 - 5.7	2.9 - 1.3	3.5-39.
Φ1/Φ2	0.3 - 21.	1.7 - 23.	2.4 - 50.	1.3-12.	1.6 - 12.	1.2-20.
D20 (%)	099.34	081.96	089.14			
Boron (PPM)					03400.	
k	1.003	1.009	0.998	1.006	1.003	1.018
	<u>+0.012</u>	<u>+0.016</u>	<u>+</u> 0.006	<u>+</u> 0.011	<u>+0.012</u>	+0.014
# of cases	16	16	26	82	63	42

parameters by few group diffusion theory. In this research, only the infinite lattice calculations were utilized. The relevant unit cell geometry is shown in Figure 2.3. The difference between this method of calculation and the LEOPA? methodology is the inclusion of spatial transport calculations to treat the unit cell geometry. In addition, in the HAMMER system most heavy metal isotopes are self-shielded whereas LEOPARD only self-shields the most abundant heavy isotope. This is quite important, for example, in the analysis of mixed thorium-uranium unit cell configurations.

Two components of the HAMMER program library are of particular interest. The THERMOS program (H-2) performs a multigroup calculation of the thermal flux distribution $(E_n^{<}.625 \text{ eV})$ by using integral transport theory. Unlike LEOPARD, the assumption of space-energy separability is not made. THERMOS uses an integral form of the transport equation to calculate the spatially dependent thermal neutron spectrum in a cell characterized by one-dimensional symmetry (D-3).

The HAMLET program performs a multigroup spatial calculation of the flux in the energy range from 0.625 eV to 10 MeV. Collision probabilities are calculated under the assumption of cosine currents crossing region boundaries and the energy spectrum during moderation is calculated as in the MUFT code.

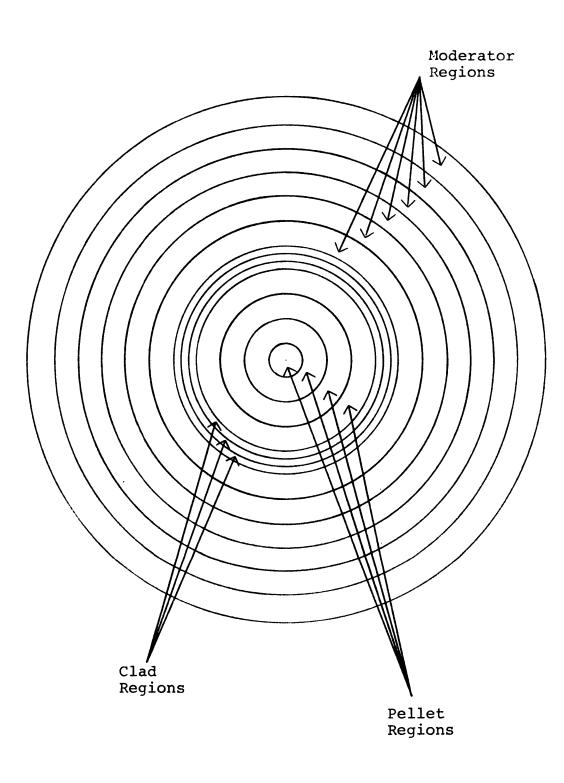


FIG. 2.3 HAMMER UNIT CELL GEOMETRY

2.5 PDQ-7, HARMONY

The PDQ-7 code (C-3, H-3) is designed to perform the spatial diffusion calculation in one, two, or three dimensions. It has been used widely by the nuclear industry for the licensing of reload cores (A-4, A-1). In this research, two energy groups are used, whereas, as many as five energy groups can be handled by the program. Solutions to the eigenvalue equation can be obtained in rectangular, cylindrical, spherical, or hexagonal geometry. Zero flux or zero current boundary conditions are admissible.

The multigroup diffusion equation is solved by discretizing the energy variable and finite-differencing the spatial variable over a variable or constant mesh size.

The one dimensional group equations are solved by Gauss elimination, and the two dimensional equations are solved by using a single-line cyclic Chebyshev semi-iterative technique. A block Gauss-Seidel procedure is used in three dimensions. In the present work, however, three dimensional PDQ-7 analyses were not performed.

The actual depletion chain manipulation and crosssection data management is performed by the HARMONY system (B-5). The depletion chains are user inputs and the crosssection tables are obtained from the LEOPARD code after appropriate massaging by the CHIMP code.

The fluxes are normalized based on the power level input by the user. The cross-sections are fitted to the

concentrations of the isotopes with time via a high order polynomial. At the end of each depletion time step, the PDQ-7 code solves for the spatial flux shape and these values of the flux levels are used in the following time step. The depletion can be done on a "point" or "block" basis. Various options available in the code and used in this research will be discussed as the need arises.

2.6 Chapter Summary

In this chapter the computer methods used in the present research have been outlined. An accurate determination of the neutron spectrum and the spatial flux shape is essential for the determination of uranium consumption. The suggested improvements in core design and fuel management which have to be evaluated involve perturbations of the neutron spectrum, and the arrangement of fuel in patterns which generate flux shapes quite different from those encountered in current designs. Hence we will routinely fall back upon state-of-the art, well-benchmarked methods to provide a solid base upon which to build further analyses.

CHAPTER 3

A POWER WEIGHTING ALGORITHM FOR ESTIMATION OF SYSTEM REACTIVITY

3.1 Introduction

First approximations to nuclear fuel management problems are often made using models in which either reactivity, ρ , (or multiplication factor, k) vary linearl (or quadratically) with burnup [G-3, N-3, S-6, S-7].

In this chapter a pragmatic case will be made for the use of ρ versus burnup as opposed to k_{∞} versus burnup. An analysis will be carried out to demonstrate that the reactivity of fuel bundles in nuclear reactors should be weighted not merely by their mass or volume fractions, but rather by their power fractions. The linear reactivity model is extended to include these effects, and computergenerated results are analyzed to verify the analytical models.

Further extension of the linear reactivity model is made to account for core leakage. This extension is also checked and found to be consistent with more detailed computer studies. Finally, an illustrative example is presented, showing the application of the model to discern the differences between uranium utilization in the "out-in" and "low-leakage" fuel management schemes.

3.2 The Linear Reactivity Model

The time-dependent evolution of the reactor or region-wise eigenvalue is an extremely useful integral measure of the neutronic status of the reactor fuel. Many researchers in the field of PWR fuel management have made good use of the empirical observation that the plot of eigenvalue versus burnup is approximately linear [G-3, N-3].

It is not widely recognized, however, that reactivity is a considerably more linear function of burnup than the eigenvalue (or k), that is:

$$\rho_{\infty} = \frac{k_{\infty} - 1}{k_{\infty}} = \rho_{0} - AB \qquad (3.1)$$

The Maine Yankee supercell reference lattice (see Appendix A) was analyzed using LEOPARD, for enrichment variations ranging from 1.5 w/o U-235 to 4.34 w/o U-235. Linear least squares curve-fits of k_{∞} versus burnup over the range 150-50,000 MWD/MT gave an average coefficient of determination (R²) of 0.9898, whereas linear fits of ρ_{∞} versus burnup gave an average R² value of 0.995 (i.e., in the latter case only 0.5% of the variance is not accounted for by the correlation!)

In a one-batch PWR core the excess reactivity at the beginning of the cycle is balanced by the presence

of the soluble boron control absorber in the moderator. As the excess reactivity of the core decreases due to fuel depletion and fission product buildup, the boron is slowly withdrawn, until at end-of-cycle the excess reactivity of the core medium is equal to the fraction of neutrons that are lost to leakage. Plots of k_{∞} and c_{∞} vs. burnup for a 3.0 w/o U-235 lattice (Maine Yankee supercell) are shown in Fig. 3.1.

In the simplest version of the linear reactivity model (leakage effects not included) the reactivity, p, decreases linearly with burnup, and for one-batch irradiation the EOC burnup is given by:

$$B_1 = \frac{c_0}{A} \tag{3.2}$$

Typical values for a 3.0 w/o U-235 enriched PWR lattice are:

$$\rho_0 \simeq 0.20$$

A $\simeq 0.91 \times 10^{-5}$ (MWD/MT)⁻¹

 $B_1 \simeq 22,000 \text{ MWD/MT}$

In an N-batch steady-state core, at the end of the reactor cycle the freshest batch is burned to $B_{discharge}/N$, the next oldest batch to $2 B_{discharge}/N$, etc. The mean reactivity of the mixture is typically estimated by simple arithmetic averaging of the EOC reactivity of each batch:

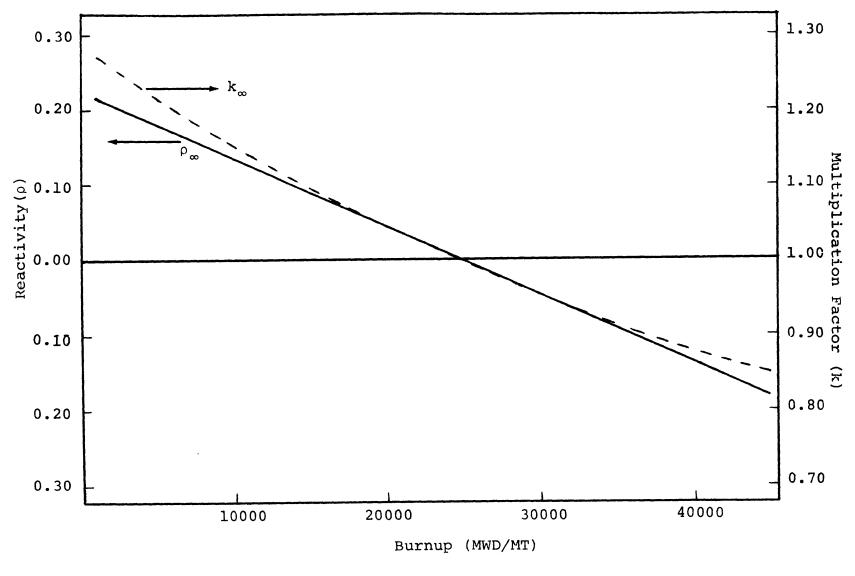


FIG. 3.1 k AND ρ_{∞} AS A FUNCTION OF BURNUP FOR A 3.0 W/O U-235 MÄINE YANKEE SUPERCELL

$$\rho_{1} = \rho_{O,N} - \frac{1}{N} \stackrel{AB}{}_{dis}$$

$$\rho_{2} = \rho_{O,N} - \frac{2}{N} \stackrel{AB}{}_{dis}$$

$$\cdot$$

$$\cdot$$

$$\rho_{N} = \rho_{O,N} - AB_{dis}$$

$$\rho_{i} = \frac{1}{N} \sum_{i=1}^{N} \rho_{i} = \frac{1}{N} \left\{ N \rho_{O,N} - \frac{N(N+1)}{2N} AB_{dis} \right\} = 0 \quad (3.3)$$

Thus, the reload reactivity needed to achieve a discharge burnup of B_1 is:

$$\rho_{0,1} = \frac{N+1}{2N} AB_1 = \frac{N+1}{2N} \rho_{0,n}$$
 (3.4)

If, instead, the reload reactivity is kept the same, then a higher burnup can be achieved:

$$B_{N} = (\frac{2N}{N+1}) B_{1}$$
 (3.5)

When the oldest batch is replaced by fresh fuel we can repeat the averaging process to find the BOC reactivity:

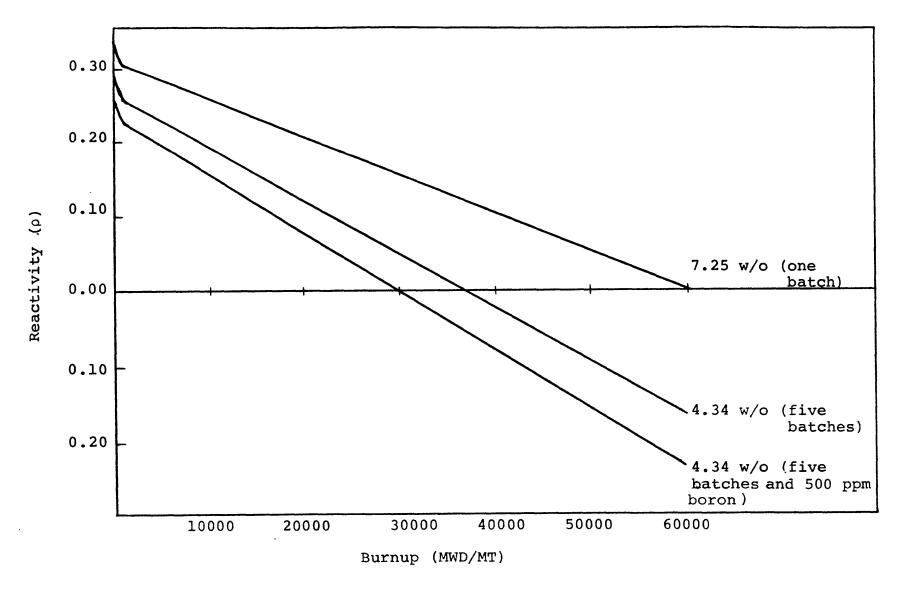


FIG. 3.2 REACTIVITY VS. BURNUP CURVES FOR FIVE BATCH FUEL MANAGEMENT SCHEME WITH 60000 MWD/MT TARGET BURNUP

$$\rho_{1} = \rho_{0,N}$$

$$\rho_{2} = \rho_{0,N} - \frac{AB_{dis}}{N}$$

$$\cdot$$

$$\cdot$$

$$\rho_{N} = \rho_{0,N} - AB_{dis}$$

$$\rho = \frac{1}{N} \sum_{i=1}^{N} \rho_{i} = \frac{1}{N} \left\{ N \rho_{0,N} - AB_{dis} \left(\frac{N(N+1)}{2N} - 1 \right) \right\} \quad (3.6)$$
Hence the cycle reactivity swing $(\rho_{0,N} - \rho_{0,N})$ is given

Hence the cycle reactivity swing ($\rho_{BOC} - \rho_{EOC}$) is given by:

$$\Delta \rho_{C,N} = \frac{1}{N} \rho_{O,1} = \frac{2}{N+1} \rho_{O,N}$$
(3.7)

Figure 3.2 illustrates an application of the linear reactivity model (LRM) methodology. A 7.0 w/o U-235 enriched PWR lattice is depleted until $\rho = 0$ at 60000 MWD/MT, the target burnup for one-batch fuel management. The reload reactivity is then determined using Eq. 3.4 such that the new reload reactivity corresponds to the anticipated use of five fuel batches. If the slopes of both curves were the same, the required enrichment would be easily calculated; however, the lower enrichment cores have a steeper ρ versus B slope, so that a higher than might be anticipated enrichment (here 4.34 w/o U-235) is required to reach criticality at EOC with the fifth and final batch at a discharge burnup of 60000 MWD/MT. In the third curve, a fixed amount of boron is added to the lattice such that $\int_{0}^{B} dis \rho(B) dB = 0$, i.e, the net excess of neutrons above the $\rho = 0$ line is balanced by the net deficit of neutrons below the $\rho = 0$ line. The fixed amount of boron is the reactor cycle-average boron concentration. Note in the figure that except for the sudden drop of reactivity during the first 150 MWD/MT (on account of the buildup of xenon and samarium fission product poisons to their equilibrium levels) the curves are very linear over the full range of burnup.

Several implicit assumptions are present in the linear reactivity model, as usually applied, which restricts its usefulness for the purpose of this research; namely, fixed batch size, equal power sharing among batches, failure to account for leakage from the core, and treating the burnup slope as a constant over the entire range of burnup. A relaxation of these conditions is clearly desirable.

3.3 Power Weighting

In a large reactor with negligible leakage and N fuel regions, the eigenvalue equation can be written as:

$$k_{sys} = \frac{\nu \Sigma_{f}^{(1)} \phi^{(1)} + \nu \Sigma_{f}^{(2)} \phi^{(2)} + \dots + \nu \Sigma_{f}^{(N)} \phi^{(N)}}{\Sigma_{a}^{(1)} \phi^{(1)} + \Sigma_{a}^{(2)} \phi^{(2)} + \dots + \Sigma_{a}^{(N)} \phi^{(N)}}$$
(3.6)

where $v\Sigma_{f}^{(i)}$ and $\Sigma_{a}^{(i)}$ are the neutron production and neutron destruction cross-sections in the ith region, respectively. This equation can be rewritten:

$$\mathbf{k}_{sys} = \frac{\nu \Sigma_{f}^{(l)} \phi^{(l)} + \nu \Sigma_{f}^{(l)} \phi^{(l)} + \cdots + \nu \Sigma_{f}^{(N)} \phi^{(N)}}{\frac{\nu \Sigma_{f}^{(l)} \phi^{(l)}}{\mathbf{k}_{\infty}^{(l)}} + \frac{\nu \Sigma_{f}^{(l)} \phi^{(l)}}{\mathbf{k}_{\infty}^{(l)}} + \cdots + \frac{\nu \Sigma_{f}^{(N)} \phi^{(N)}}{\mathbf{k}_{\infty}^{(N)}}$$
(3.9)

Rearranging the terms gives:

$$\frac{1}{k_{\rm sys}} = \frac{q^{(1)}}{k_{\infty}^{(1)}} + \frac{q^{(2)}}{k_{\infty}^{(2)}} + \dots + \frac{q^{(N)}}{k_{\infty}^{(N)}}$$
(3.10)

where $k_{\infty}^{(i)}$ is the infinite medium multiplication factor of the ith region and q⁽ⁱ⁾ is the fission neutron production fraction in region i:

$$q^{(i)} = \frac{\nabla \Sigma_{f}^{(i)} \phi^{(i)}}{\sum_{i=1}^{N} \nabla \Sigma_{f}^{(i)} \phi^{(i)}}$$
(3.11)

In a pressurized water reactor, the value of κ/ν (the ratio of the average energy released per fission to the average number of neutrons released per fission) varies slightly with burnup, as shown in Fig. 3.3. Nevertheless

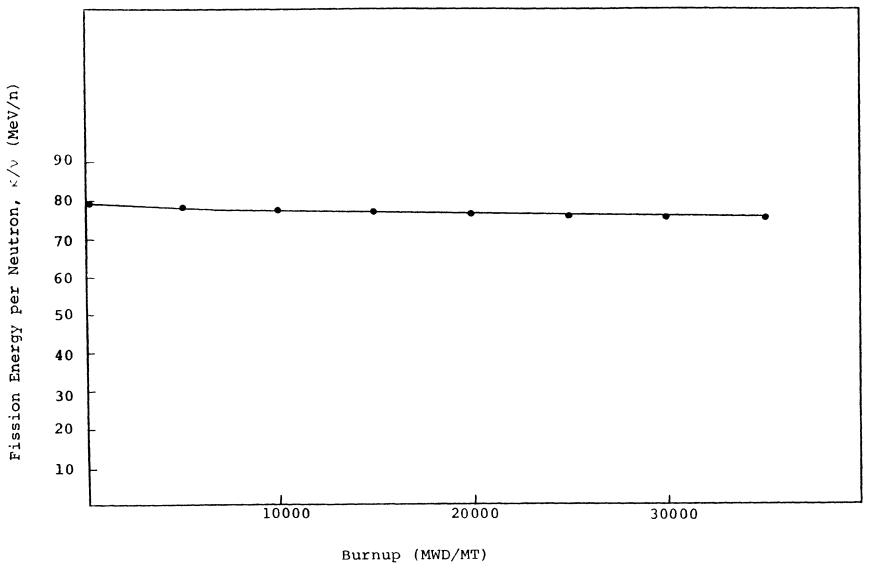
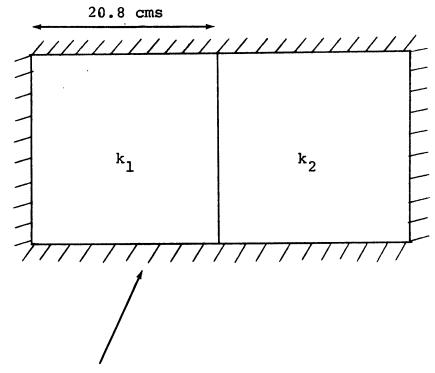


FIG. 3.3 AVERAGE ENERGY RELEASE PER FISSION NEUTRON (κ/ν) As a function of fuel burnup

TABLE 3.1	COMPARISON OF COMPUTED (PDQ-7) AND POWER-WEIGHTED
	VALUES OF k FOR A TWO-BUNDLE PROBLEM (FIG. 3.4)

۲.

		2		
k	k	$\sum_{i=1}^{f_i \rho_i}$	k _∞ (PDQ-7)	Δ k
1.128962	1.073943	1.103352	1.103167	0.000185
1.128962	1.000420	1.075070	1.074255	0.00081
1.128962	0.900117	1.046842	1.044636	0.00220
1.128962	0.805563	1.028687	1.024873	0.003814
1.073943	0.805563	0.985596	0.983145	0.002451
1.073943	1.000420	1.040755	1.040527	0.000228
0.900117	1.000420	0.957598	0.957212	0.000386
0.900117	0.805563	0.860263	0.859973	0.000290



Zero-Current Boundary Condition

FIG. 3.4 GEOMETRY FOR TWO-BUNDLE PDQ-7 POWER SPLIT PROBLEM

if we approximate $q^{(i)}$ by the thermal power fraction, f_i , reasonable accuracy results. Equation 3.11 then becomes:

$$\frac{1}{k_{sys}} = \sum_{i=1}^{N} \frac{f_i}{k_i}$$
(3.12)

Table 3.1 shows the results of a two-bundle, zero-current boundary condition, PDQ-7 problem where the k_{∞} values of the bundles have been varied over a wide range. The differences between the eigenvalues calculated using Eq 3.12 and the eigenvalues computed by PDQ-7 are very small. The worst case occurs when the difference between k_1 and k_2 is the largest, and even in this case, the error is less than 0.4%. Thus, the errors associated with using an average value of κ/ν do not appear prohibitive.

Since $\rho = 1 - 1/k$, Eq 3.12 can be rewritten as:

$$\rho_{\text{sys}} = \sum_{i=1}^{N} f_i \rho_i \qquad (3.13)$$

This formula indicates that the correct algorithm for reactivity combination in a PWR core is a power-weighting scheme.

The implications of power weighting become more apparent when the reactivity balance is calculated. For an N-batch steady-state core, assuming that the power fractions remain constant over a given cycle:

$$\rho_{1} = \rho_{0} - AB_{dis} f_{1}$$

$$\rho_{2} = \rho_{0} - AB_{dis} (f_{1} + f_{2})$$

$$\rho_{N} = \rho_{0} - AB_{dis} (f_{1} + f_{2} + \dots + f_{N})$$
(3.14)

At end of cycle

$$\rho = 0 = \sum_{i=1}^{N} f_{i} \rho_{i}$$
(3.15)

from which

$$\rho = 0 = \rho_{o} - AB_{dis} \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} \right\}$$
(3.16)

r

or

$$B_{dis} = \frac{\rho_{o}}{A\left\{\sum_{i=1}^{N}\sum_{j=i}^{N}f_{i}f_{j}\right\}}$$
(3.17)

The double summation in the denominator, which we will designate as the "cycle schedule index", accounts for the effect of burnup schedule (the sequence of f_i), or alternatively, the effect of a non-uniform core power distribution (since the f_i apply to one batch over its N in-core cycles, or all batches in-core during a representative cycle). Note that this effect is not included in the simplest version of the linear react-ivity model. As will be shown, the lower burnup associated with non-uniform f_i degrades uranium utilization by a non-negligible amount.

3.4 Accounting for Leakage

In any finite reactor the neutron leakage is an important factor in the determination of system reactivity, hence sustainable burnup and uranium utilization. In a normal PWR operated with an out-in fuel shuffling strategy, roughly 3% to 4% of the neutrons are lost via leakage to, and absorption in, the regions surrounding the core (grids, core barrel, thermal shields, etc.).

An approximate treatment of leakage effects can be developed by considering the fast group leakage as a function of source shape. The plane geometry (or large radius cylinder) flux kernel (flux at x due to a source at x') is given by:

$$\phi(x, x') = \frac{M}{2D} \exp\left\{-\frac{(x'-x)}{M}\right\}$$
 (3.18)

and the neutron current can be written as:

$$J = -D \frac{d\phi}{dx} = \frac{1}{2} \exp \left\{ -\frac{(x'-x)}{M} \right\}$$
(3.19)

Consider a fuel region of total length 2H with a symmetric source shape about the origin. The fraction of neutrons which leak out the end of the region (the leakage reactivity loss) is: $\int_{H}^{H} \int_{H}^{H} \int_{H}^{$

$$\rho_{\rm L} = \frac{\int_{0}^{\pi} J(x,x') S(x) dx}{\int_{0}^{H} S(x) dx} = \frac{\int_{0}^{\pi} \frac{1}{2} \exp\left[-\frac{(\pi - x)}{M}\right] S(x) dx}{\int_{0}^{H} S(x) dx}$$
(3.20)

There are three source shapes which are of immediate interest with respect to the behavior of ρ_L as a function of source shape; namely, flat (S(x) = S₀), cosine (S(x) = S₀ cos $\frac{\pi x}{2H}$), and flat interior with drooping ends (S(x) = S₀(1-exp - $\frac{(H-x)}{M}$). The corresponding solutions of the leakage reactivity equations are shown in Table 3.2.

In a PWR the source shape is cosine in the axial direction at beginning of life and flat with drooping ends at end of life. Examination of the kernel equations indicates that the leakage effect is most prominent in the last few diffusion lengths of the fuel material, that is, most of the neutron loss originates in fuel regions within two or three migration lengths of the core periphery. Since this is the case, axial and radial leakage should be correlated with the power (source) in the peripheral core regions. Figure 3.5 shows the axial leakage reactivity loss plotted as a function of the power in the outermost 6 inches of a Maine Yankee fuel assembly in which the last six inches of the fuel have been replaced with various enrichments from 0.2 w/o U-235 to 1.0 w/o U-235, and with depleted and natural uranium fuel blankets, at various stages of assembly burnup. The calculations were made in two groups using PDQ-7 and PDQ-7-HARMONY. The correlation is linear, with an R² value of 0.98445. The non-zero intercept can be interpreted as a relative indication of leakage due to neutrons born in the interior.

TABLE 3.2 EFFECTS OF SOURCE SHAPE ON LEAKAGE REACTIVITY

SOURCE (POWER) SHAPE	EQUATION FOR SOURCE	EQUATION FOR ρ_{L}	LEAKAGE REACTIVITY
FLAT	S(x) = 1.0	$\frac{\int_{0}^{H} \frac{1/2 \left[\exp\left(\frac{H-x}{M}\right)\right] dx}{\int_{0}^{H} dx}$	0.02135
COSINE	$S(\mathbf{x}) = S_0 \cos(\frac{\Pi \mathbf{x}}{2H})$	$\frac{\int_{0}^{H} \frac{1/2 \left[\exp(\frac{H-x}{M}) \right] \cos \frac{\pi x}{2H} dx}{\int_{0}^{H} \cos \frac{\pi x}{2H} dx}$	0.00223
FLAT WITH DROOPING ENDS	$S(\mathbf{x}) = S_0(1-\exp(\frac{H-\mathbf{x}}{M}))$	$\frac{\int_{0}^{H} \frac{1/2 \left[\exp\left(\frac{H-x}{M}\right) \right] - \frac{1}{2} \left[\exp\left(\frac{H-x}{M}\right) \right]}{\int_{0}^{H} 1 - \exp\left(\frac{H-x}{M}\right) dx}$)dx 0.01613

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* For representative parameter values, M=7.5 cms, H=175.2 cms.

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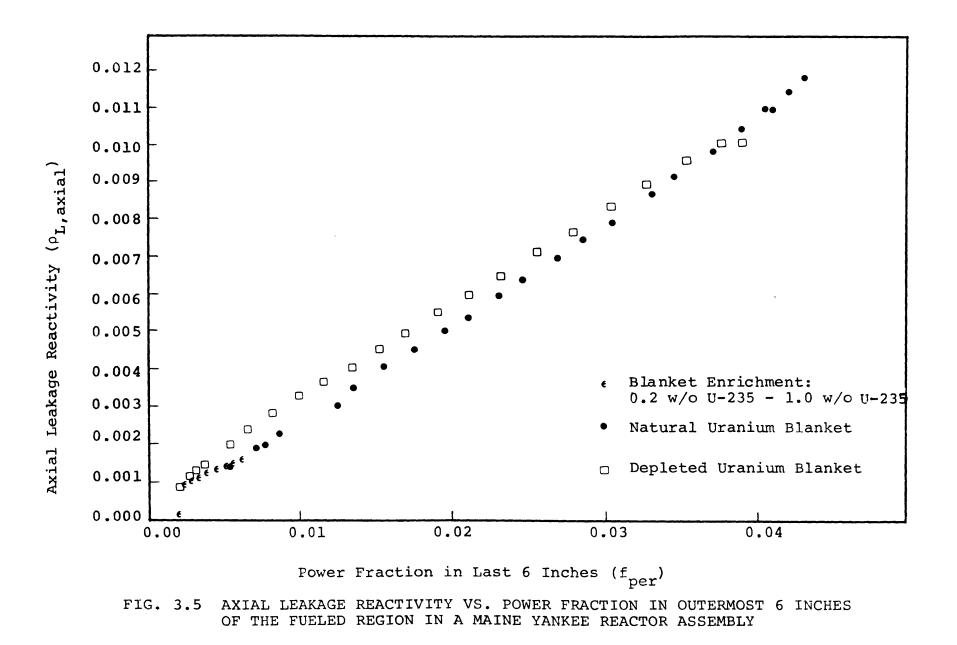


Figure 3.6 shows the <u>radial</u> leakage reactivity loss as a function of the power in the peripheral fuel assemblies of the Maine Yankee reactor, again using PDQ-7 static calculations. The peripheral power was varied by successively shuffling the four fuel batches in the core between the periphery and the interior. The lowest power case represents 4th cycle fuel on the core periphery, whereas the highest power case represents fresh fuel loaded on the core periphery. The absorption fraction in nonfuel, ex-core materials was used to measure the leakage reactivity.

For the comparison of various fuel management schemes, the axial and radial leakage will be assumed to be "decoupled", i.e., changes in radial leakage will not affect the axial losses and vice versa. As shown, both leakage components can be correlated as linear functions, where:

$$\rho_{\rm L} = \alpha + \beta f_{\rm per} \tag{3.21}$$

The values of α and β depend on the specific reactor design; f_{per} is the fraction of the core power generated in the peripheral fuel region.

System reactivity can be defined as:

$$\rho_{\text{sys}} = \frac{\left\{ \sum_{i=1}^{N} (F_{i} - A_{i}) \right\} - A_{R}}{\sum_{i=1}^{N} F_{i}} = \left\{ \sum_{i=1}^{N} f_{i} \rho_{i} \right\} - \frac{A_{R}}{\sum_{i=1}^{N} F_{i}}$$

$$(2.2)$$

(3.22)

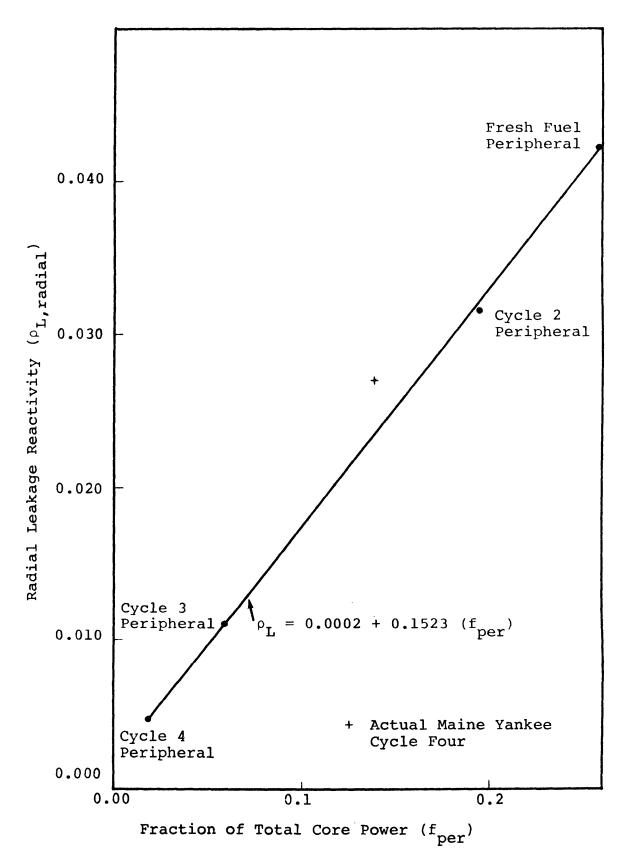


FIG. 3.6 RADIAL LEAKAGE REACTIVITY FOR M.Y. REACTOR VS. FRAC-TION OF CORE POWER GENERATED IN PERIPHERAL ASSEMBLIES

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where ρ_i is the assembly reactivity, $\frac{(F_i - A_i)}{(F_i)}$, F_i is the neutron production by fission in fuel region i, A_i is the neutron loss by absorption in fuel region i and A_R is the absorption in the non-fuel, ex-core regions. The term $\frac{A_R}{\sum_{i=1}^{N} F_i}$ can be identified as the leakage reactivity.

Thus,

$$\rho_{\text{sys}} = \left\{ \sum_{i=1}^{N} f_{i} \rho_{i} \right\} - \rho_{L}$$
(3.23)

and the discharge burnup can be calculated as:

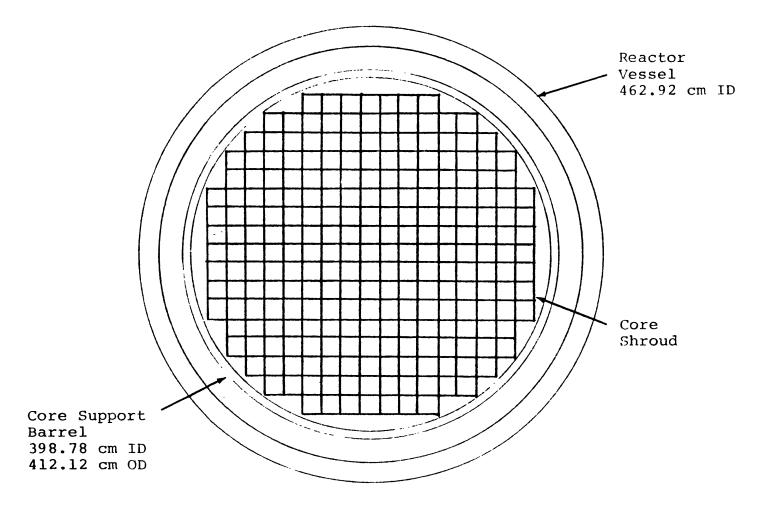
$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A\left\{\sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j}\right\}}$$
(3.24)

If the reactivity decrement associated with axial leakage is combined with the fuel reactivity:

$$\rho'_{O} = \rho_{O} - \rho_{L,axial}$$
(3.25)

then the radial leakage correlation can be included in Eq. 3.24 to give:

$$B_{\text{discharge}} = \frac{\rho_{O}' - \alpha - \beta f_{\text{per}}}{A\left\{\sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j}\right\}}$$
(3.26)



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FIG. 3.7 C-E SYSTEM 80TM CORE DESIGN (241 FUEL ASSEMBLIES)

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3.5 An Application of The Advanced Linear Reactivity Model: Low Leakage Fuel Mangement

One of the benefits of the Advanced Linear Reactivity Model (ALRM) is its ability to calculate the differences in uranium utilization between fuel management schemes where the application of simpler linear reactivity models fails to discern any difference. In addition, the competing effects of leakage and fuel power history factors can be separated and quantified.

Consider five-batch, extended burnup, "out-in" fuel management versus a comparable "low-leakage" burnup schedule. The Combustion Engineering System-80 reference design shown in Fig. 3.7 will be used as the core model. The reactor has 241 assemblies with 48 peripheral assembly locations. Published data (M-3) for low-leakage and out-in fuel management allows us to correlate leakage reactivity with peripheral power fraction, as the core approaches its equilibrium cycle, as shown in Fig. 3.8.

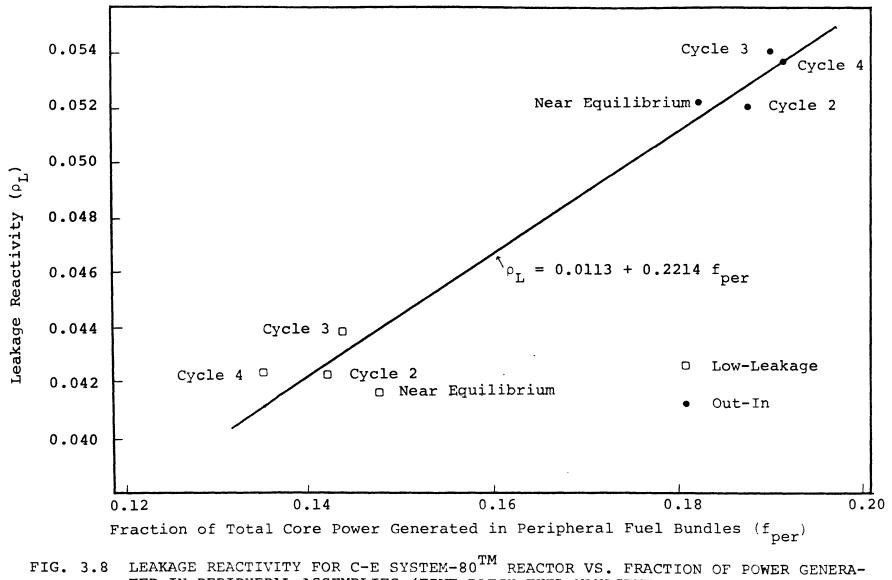
$$\rho_{\rm L} = 0.01126 + 0.2214 \, \rm f_{per} \tag{3.27}$$

The power fractions for each of the five batches under "equilibrium" cycle conditions are shown in Table 3.2.

Typical lattice parameters for 5-batch, extended burnup fuel management are:

$$\rho_{o} \simeq 0.2661$$

A $\simeq 0.7154 \times 10^{-5} (MWD/MT)^{-1}$



TED IN PERIPHERAL ASSEMBLIES (FIVE BATCH FUEL MANAGEMENT, HIGH BURNUP)

The equation for the discharge burnup becomes:

$$B_{dis} = \frac{0.2661 - 0.01126 - 0.2214 f_{per}}{(0.7154 \times 10^{-5}) \left\{ \sum_{i=1}^{5} \sum_{j=i}^{5} f_{i} f_{j} \right\}}$$
(3.28)

For the power fractions listed in Table 3.3, the lowleakage fuel arrangement achieves an average discharge burnup of 51542 MWD/MT, whereas the out-in fuel management scheme achieves a burnup of 49790 MWD/MT; a net advantage to low leakage management under equilibrium cycle conditions of 3.52%. This value neglects the presence of residual shim worth associated with the use of burnable poison to suppress the high assembly power peaking factors encountered in the low-leakage fuel arrangements. Under equilibrium conditions it is predicted (M-3) that the shim residual worth would drop to 0.5% $\Delta \rho$. Equation 3.28 then becomes:

$$B_{dis} = \frac{0.2661 - 0.01126 - 0.2214 f_{per} - \Delta \rho_{res}}{(0.7154 \times 10^{-3}) \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} \right\}}$$
(3.29)

where $\Delta \rho_{res} = +0.005$. Under these conditions the discharge burnup becomes 50381 MWD/MT for the low-leakage scheme and the relative advantage in discharge burnup, hence uranium utilization is 1.2%. The equilibrium cycle uranium savings predicted by CE using detailed computer calculations is 2.34%. The difference is caused by the use of a leakage

TABLE 3.3	EQUILIBRIUM CYCLE POWER FRACTIONS FOR 5 BATCH
	OUT-IN AND LOW-LEAKAGE FUEL MANAGEMENT SCHEMES*

	POWER FRA	ACTION
BATCH NUMBER	<u>OUT-IN</u>	LOW-LEAKAGE
1	0.1860**	0.2514
2	0.2400	0.1612**
3	0.2140	0.2063
4	0.1960	0.1971
5	0.1640	0.1840
PERIPHERAL	0 1001	0.1400
POWER FRACTION	0.1831	0.1480

WER	FRACTION	0.1831	0.1480

* Ref. (M-3).

** Principal Peripheral Batch.

correlation based on an approach to equilibrium cycle whereas the CE results are based on achieving the equilibrium cycle leakage profiles. In view of the simplicity of the ALRM, the agreement must be considered good; and it could probably be improved upon with experience in interfacing the ALRM and the results of computer modeling.

For both fuel management schemes, the

burnup schedule index $\sum_{i=1}^{N} \sum_{j=i}^{N} f_i f_j$ is approximately 0.602. (The minimum burnup schedule index occurs when all f_i are equal (see Appendix C). Under these conditions, the schedule index becomes 0.600, and the discharge burnup for a perfectly power flattened core becomes 49054 MWD/MT. This represents a 1.5% disadvantage vs. the conventional out-in fuel management strategy and a 3.8% disadvantage vs. the low-leakage scheme, because of the much higher radial leakage.) This clearly demonstrates the importance of the leakage and cycle schedule index in uranium utilization calculations, even though such perfectly flat distributions probably cannot be attained in a real reactor, at least not without encountering compensatory losses from residual burnable poison.

3.6 Chapter Summary

In this chapter, the linear reactivity model has been extended to include the effects of inter-batch power sharing and neutron leakage losses from the core. Its

application to a detailed design study indicates reasonably good agreement for such a simple model, and serves to highlight the importance of the cycle schedule index and leakage in burnup calculations. While useful in principle, the ALRM, as presented so far, is restricted in practice by the need for sufficiently accurate specification of the batch power sharing fractions, f_i. If detailed PDQ computations were needed to supply this information one might question whether much was accomplished in then applying a simple model. Thus we must also stress the need for developing a sufficiently accurate but simply implemented power sharing algorithm. This task will be addressed in the next chapter, where other restrictions on the model will also be relaxed.

CHAPTER 4

A POWER SHARING ALGORITHM FOR PWR FUEL BUNDLES

4.1 Introduction

In the previous chapter, the influence of the cycle schedule index on uranium utilization was demonstrated. The power fractions used in the calculations were extracted from detailed computer results. In this chapter, an empirical relation will be proposed to describe the power sharing among in-core batches in a PWR. This will provide the remaining ingredient necessary for a simple model which can stand on its own.

Using a "group-and-one-half" model, and the assumption that the fast flux in a bundle is proportional to $k_{\infty}^{-\theta-1}$, leads to a prescription for bundle power as a function only of the k_{∞} values of the bundles involved.

In this chapter two and nine-bundle, zero-current boundary condition problems are analyzed with PDQ-7, and θ values are calculated and compared. Then quarter-core results are analyzed and a heuristic algorithm for batch power fractions is developed. Finally, the linearity condition on the ρ vs B curve is relaxed, and a computer program is described which calculates equilibrium cycle fuel discharge burnups, including the effects of leakage, variable batch size, and cycle schedule index. 4.2 "Group and One-Half" Model

In the two-group formulation, the thermal power in a region can be written as:

$$q = \kappa \Sigma_{f1} \phi_1 + \kappa \Sigma_{f2} \phi_2 \tag{4.1}$$

If the assumption $\Sigma_{12} \phi_1 = \Sigma_{a2} \phi_2$ is made, and the leakage effects are neglected, where Σ_{12} is the macroscopic down-scatter cross-section from group one (the fast group) to group two (the thermal group), Eq. 4.1 becomes:

$$q = \kappa \left\{ \Sigma_{fl} \phi_{l} + \Sigma_{f2} \frac{\Sigma_{12}}{\Sigma_{a2}} \phi_{l} \right\}$$
(4.2)

The two-group value of k_{m} can be written as:

$$k_{\infty} = \frac{\nu \Sigma_{f1}}{\Sigma_{a1} + \Sigma_{12}} + \frac{\Sigma_{12}}{\Sigma_{a1} + \Sigma_{12}} \frac{\nu \Sigma_{f2}}{\Sigma_{a2}}$$
(4.3)

Thus, the power can be written as:

$$q = \frac{\kappa}{\nu} (\Sigma_{a1} + \Sigma_{12}) k_{\infty} \phi_1 \qquad (4.4)$$

If the quantity $\underline{\kappa}(\Sigma_a + \Sigma_{12})$ is assumed to be constant, and the empirical relation $\phi_1 \simeq k_{\infty}^{\theta-1}$ is invoked (see Appendix D), a formula for q can be proposed as:

$$q = Ck_{\infty}^{\theta}$$
 (4.5)

4.3 Results of Two-Bundle Calculations

Consider the two-bundle, zero-current boundary condition problems discussed in Section 3.3. If the power in bundle one is Ck_1^{N} and the power in bundle two is Ck_2^{N} , then the power weighting algorithm gives a combined system reactivity of:

$$k_{sys} = \frac{k_1^{M} + k_2^{N}}{k_1^{M-1} + k_2^{N-1}}$$
(4.6)

It can be shown (see Appendix D) that for such a system a single number, G, can be found such that:

$$k_{sys} = \frac{k_1^{G} + k_2^{G}}{k_1^{G-1} + k_2^{G-1}}$$
(4.7)

In this case, the power fraction can be approximated as:

$$f_1 = \frac{k_1^G}{k_1^G + k_2^G}$$
(4.8)

Table 4.1 shows the results of a series of two-bundle, zero-current boundary condition problems. The burnup of the bundles was varied to provide a number of test cases with different values of k_1 and k_2 . In all cases, the G

		PDQ RES	SULTS		G =	3.6
k_1	k ₂	fl	f ₂	G	f ₁	f ₂
1.3117	1.2094	0.5754	0.4246	3.75	0.5726	0.4274
1.3117	0.9638	0.7449	0.2550	3.48	0.7520	0.2480
1.1289	1.0739	0.5469	0.4531	3.77	0.5448	0.4552
1.1289	1.0004	0.6099	0.3901	3.70	0.6071	0.3929
1.1289	0.9001	0.6915	0.3085	3.56	0.6933	0.3067
1.1289	0.8056	0.7572	0.2428	3.37	0.7711	0.2289
1.0739	0.8056	0.7309	0.2691	3.48	0.7379	0.2621
1.0739	1.0004	0.5661	0.4339	3.75	0.5635	0.4365
0.9001	1.0004	0.4013	0.5987	3.79	0.4060	0.5940
0.9001	0.8056	0.6053	0.3947	3.85	0.5985	0.4015

TABLE 4.1 ACTUAL AND ESTIMATED POWER FRACTIONS FOR TWO BUNDLE,* ZERO CURRENT BOUNDARY CONDITION PROBLEMS

* Configuration: Fig. 3.4.

factor was calculated which reproduced the exact (PDQ-7 computed) answer. An average G factor of 3.6 was then adopted for all cases and the errors associated with this assumption determined.

As shown in the table, the errors associated with assuming an "average" value of G of 3.6 are very small: averaging less than 1.3%.

It can also be seen that the larger the difference $(k_1 - k_2)$, the smaller the G value becomes. The largest errors in estimating the power fractions occur when the actual G value is farthest from the average value of G = 3.6.

In a power-flattened PWR core, high k_{∞} bundles are surrounded by low k_{∞} bundles in a well shuffled pattern. To determine a θ value more representative of such cores, the configuration and boundary conditions shown in Fig. 4.1 were used in a second series of PDQ-7 calculations. Three successive constant values of k_2 (0.9, 1.0, 1.2) were used and the values of k_1 were varied (as shown in Table 4.2) over a wide range of enrichments, burnups, fuel-to-moderator ratios, and boron concentrations. Fig. 4.2 shows a plot of $\ln \left\{ \frac{q_1}{q_{avq}} \right\}$ versus $\ln \left\{ \frac{k_1}{k_{sys}} \right\}$. The plot is linear, with a coefficient of determination (R²) equal to 0.975. The formula for the best least square fit is:

$$\frac{q_1}{q} = 0.94 \left\{ \frac{k_1}{k} \right\}^{1.55}$$
(4.9)

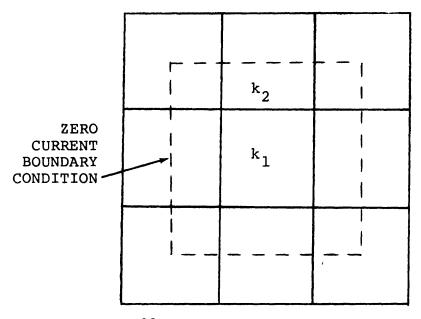




FIGURE 4.1 ASSEMBLY CONFIGURATION FOR DETERMINATION OF θ FOR pwr CORE INTERIOR CONDITIONS

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RELOAD ENRICHMENT (W/O)	BURNUP (MWD/MT)	PIN PITCH (IN.)	BORON (PPM)
4.3	150	0.580	0
4.3	10,000	0.580	0
4.3	25,000	0.580	0
4.3	40,000	0.580	0
3.0	0	0.580	200
3.0	0	0.580	400
3.0	0	0.580	600
3.0	0	0.680	800
3.0	20,000	0.580	0
3.0	30,000	0.580	0
3.0	40,000	0.580	0
4.0	0	0.580	0
3.5	0	0.580	0
2.5	0	0.580	0
2.0	0	0.580	0
3.0	0	0.500	0
3.0	0	0.575	0
3.0	0	0.600	0

TABLE	4.2	CASES A	NALYZEI) TO	DETERMINE	POWER
		SHARING	AMONG	PWR	BUNDLES	

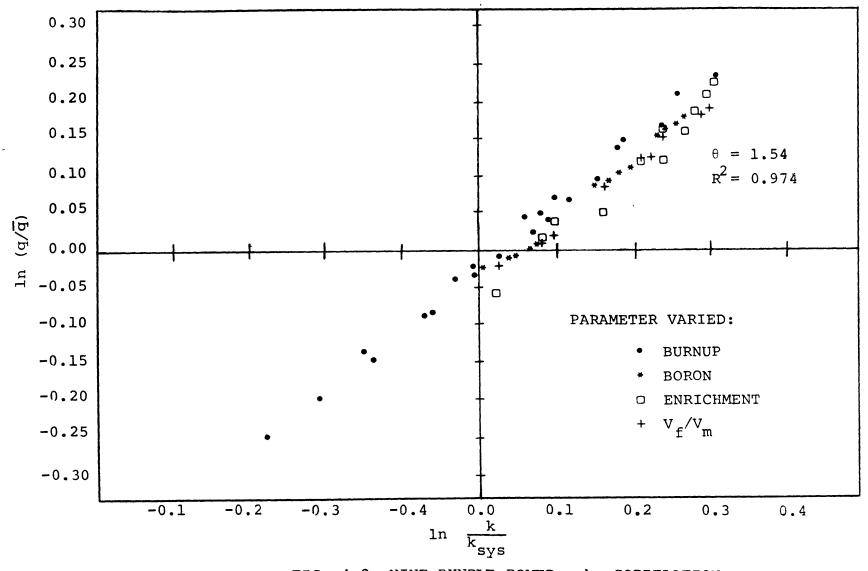


FIG. 4.2 NINE-BUNDLE POWER - $k_{_{\infty}}$ CORRELATION

4.4 Whole Core Results

The results of the simple nine-bundle zero-current boundary condition problem suggest that the power in the interior assembly can be written as:

$$q_{int} = C \left\{ \frac{k_{int}}{\overline{k}} \right\}^{\Theta} \overline{q}$$
 (4.10)

where \overline{q} represents the average power in the nine assemblies and \overline{k} represents the effective value of k for the nine assembly cluster.

The natural logarithms of interior assembly powers as a function of the natural logarithm of assembly k_{∞} values are plotted in Figs. 4.3 and 4.4 for a C-E System 80^{TM} core (W-1) and the Maine Yankee cycle-four reload design (D-4), respectively. The least squares fit for the System 80^{TM} assembly data is given by the equation

$$q_i = 1.026 k_i^{1.782}$$
 (4.11)

whereas the Maine Yankee assembly data can be fit by the equation

$$q_i = 1.003 k_i^{1.487}$$
 (4.12)

The wide scatter among the assembly data is evident from the figures. This is testified to in a quantitative sense

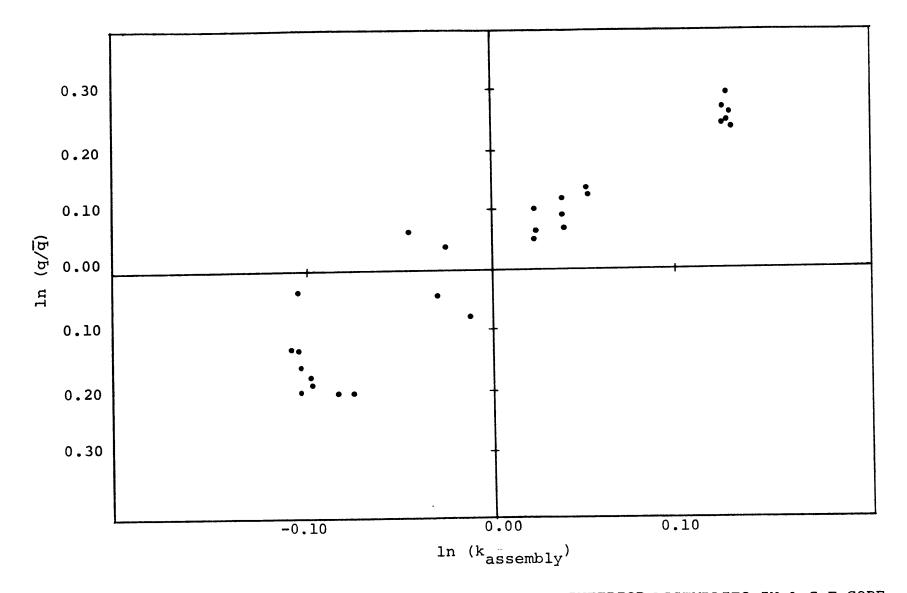
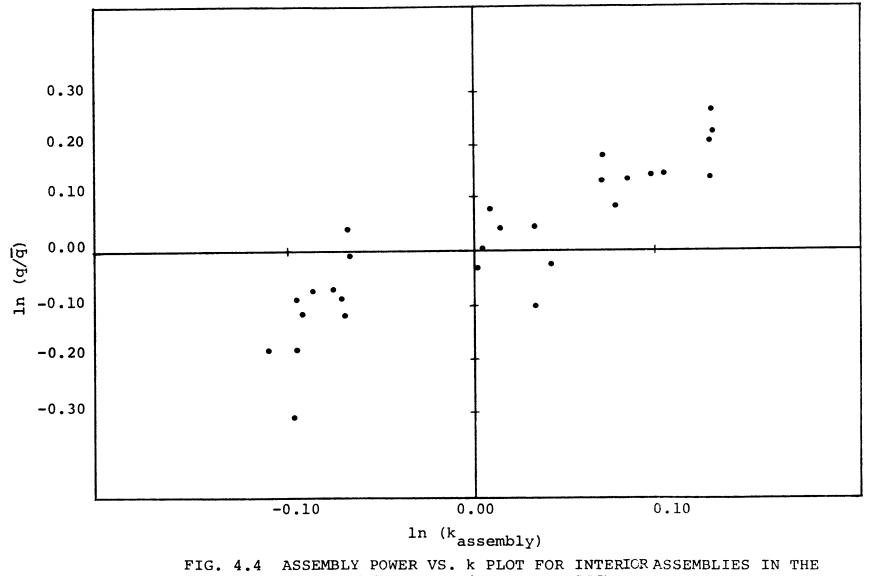
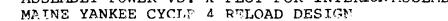


FIG. 4.3 ASSEMBLY POWER VS. k PLOT FOR INTERIOR ASSEMBLIES IN A C-E CORE





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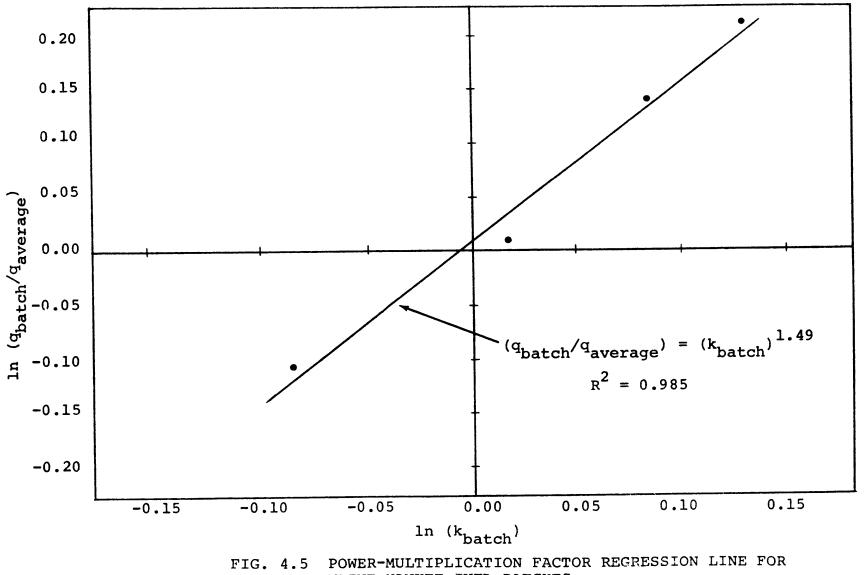
by the coefficient of determination R², which for both sets of data is in the range 0.6-0.7. However, for present purposes it is more relevant to plot the data for assemblies grouped into a batch. The results for the Maine Yankee cycle 4 reload core design are shown in Fig. 4.5. The best-fit line can be described by the equation

$$\frac{q_{\text{batch}}}{q_{\text{average}}} = 1.003 \left\{ k_{\text{batch}} \right\}^{1.49}$$
(4.13)

The coefficient of determination (R^2) for this plot is 0.985. Thus, even though individual assembly power cannot be predicted accurately, the batch-wise power splits can be determined to the level of accuracy required for detailed fuel management analysis. A similar analysis for the CE core yielded a θ value of 1.79 with an R^2 value of 0.996. The plot for the batches is shown in Fig. 4.6. These values for θ are consistent with the value of θ determined from the nine-assembly calculations, 1.55.

The core maps for the two designs analyzed in this section are shown in Figs. 4.7 and 4.8. We have, however, to this point, excluded the assemblies which are grouped on the core periphery. Estimating the powers of the assemblies grouped on the core periphery is crucial to estimation of the core leakage reactivity, and hence, the fuel discharge burnup. For the peripheral batch, the multiplication factor, k_{eff} , can be defined as:

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MAINE YANKEE FUEL BATCHES

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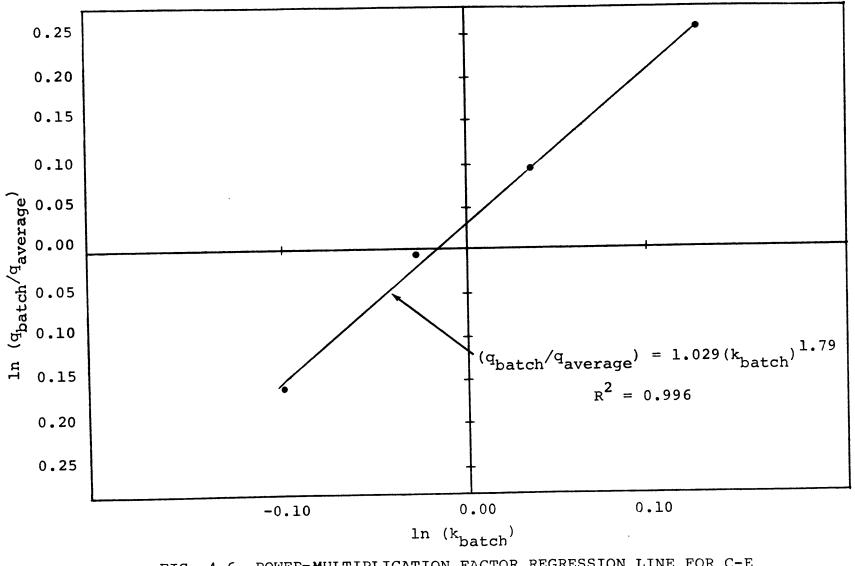


FIG. 4.6 POWER-MULTIPLICATION FACTOR REGRESSION LINE FOR C-E SYSTEM-80TM BATCHES

k q/q			,		1	0.		949
			1.194	0.698			0.952	
			0.843		0.891	[
		1.194 0.840	1		1.088 1.148		0.906 0.937	
	1.194	1.138 1.152			1.016 1.040		1.103	
	1.194 0.920	1.934 0.833					0.912	
1.194	1.195 1.087	0.928 0.931	1.003	0.917	1.110	1.908	1.082	1.006
1		1.140						1.037
1.195	0.936	1.251	0.791	1.158	0.889	1.090	0.832	0.907
0.934	1.139 1.300	0.935 1.039	1.036 1.049	1.011 1.078	1		1.036 0.905	

FIG. 4	.7	CORE	MAP	FOR	MAINE	YANKEE	CYCLE	4	REDESIGN
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k q/d	Ā						174 1. 694 0.	1
L					1.169 0.995			
					0.975 1.038			0.900 0.966
			1.135 1.306					1.139 1.267
			0.956 1.063					
	1.169 0.994		1.137 1.279					
1.174	1.168 1.121	0.901 0.854	0.919 0.815		1.055 1.132		0.901 0.860	1.069 1.158
0.694			0.905				1.054	
1.170	0.875	1.292	0.837	0.922	1.063	8.852	1.138	1.885
0.956	1.137	0.900	1.139	0.906	1.024	1.040	1.024	0.896
	1.134	0.963	1.264	0.828	1.053	1.095	1.102	0.877

FIG.	4.8	CORE CORE	MAP	FOR	COMBUSTION	ENGINEERING	SYSTEM-80 TM
		CORE					

$$k_{eff} = (PNL) k_{\infty}$$
(4.14)

where PNL is the non-leakage probability for the peripheral zone; and the peripheral zone power can be written as

$$\frac{q}{\overline{q}} = C \left\{ (PNL) k_{\infty} \right\}^{\theta} p \qquad (4.15)$$

where C is a constant and θ_p is the value of θ which is appropriate for peripheral batches. Estimation of θ_p and PNL for peripheral batches is a somewhat empirical exercise at present, but the limited data base suggests that $\theta_p \approx 2.0$ and PNL ≈ 0.77 . Thus, the power sharing formula for batch i in an N batch core with M_i assemblies in batch i is given by:

$$f_{i} = \frac{\frac{M_{i,per}}{M_{i}} (PNL_{i}k_{i})^{\theta}per + \frac{M_{i,int}}{M_{i}} (k_{i})^{\theta}int}{\sum_{i=1}^{N} \left(\frac{M_{i,per}}{M_{i}} (PNL_{i}k_{i})^{\theta}per + \frac{M_{i,int}}{M_{i}} (k_{i})^{\theta}int\right)}$$
(4.16)

where f_i is the fraction of the total core power generated in batch i, M_i is the number of assemblies in batch i, $M_{i,per}$ is the number of peripheral assemblies in batch i, $M_{i,int}$ is the number of interior assemblies in batch i, PNL is the non-leakage probability associated with peripheral assemblies in batch i, and θ_{per} and θ_{int} are, respectively, the peripheral and interior values for θ . The choice of the θ values and non-leakage probabilities requires some prior knowledge of the characteristics of the reactor core design and fuel management schemes under consideration. Equal power sharing among batches can be obtained by setting $\theta = 0.0$. This limit is, of course, not achievable in real core designs. At the other extreme an uneven power sharing schedule can be achieved by setting $\theta \simeq 3.0$. The adverse power sharing resulting from this choice leads to detrimental power peaking in the core interior.

For extreme perturbations from normal fuel management strategies, a static PDQ-7 2-dimensional calculation suffices to estimate representative θ values at low cost.

4.5 The ALARM (A-Linear Advanced Reactivity Model)

An automated procedure has been developed for determining the discharge burnup of fuel batches, in an N-batch reactor based on the θ -method described previously. The model includes features developed in Chapter 3 for the Advanced Linear Reactivity Method, such as a leakage correlation and power-weighted reactivity averaging. θ values for the core periphery and interior are input to permit estimation of a cycle schedule index. In addition, the final restriction, that of linear reactivity, is relaxed, and the \boldsymbol{k}_{∞} versus burnup values from the assembly spectrum calculation (e.g., LEOPARD) are directly input as a table. An nth order (specified by the user) Lagrangian interpolation is performed at each burnup step required in the computation. Table 4.3 summarizes the principal features of the code, and Fig. 4.9 displays a flow chart of the code. A further description of the code, a listing, and a sample problem can be found in Appendix E. The code calculates the burnups for each batch (starting with an internally generated initial estimate) and iterates until the EOC k values for batch I are equal to the BOC values of k_{m} for batch I + 1. This method is ideal for equilibrium cycle calculations. At EOC the power fractions are recalculated via the theta formula (Eq. 4-16) and the iteration is restarted until overall convergence is achieved.

Non-linear reactivity vs. burnup behavior can be accommodated

Power-weighted reactivity averaging is employed

Batch fraction need not be kept constant

Peripheral leakage is estimated using a linear reactivity/power correlation

Interior batch power is calculated using k^{θ} method

Peripheral batch power is calculated using a separate k^{θ} correlation, correcting k for leakage

Nth order Lagrangian interpolation in a $\rho(B)$ table is employed

$$\rho_{\text{sys}} = \sum_{i=1}^{N} f_{i} \rho_{i}$$

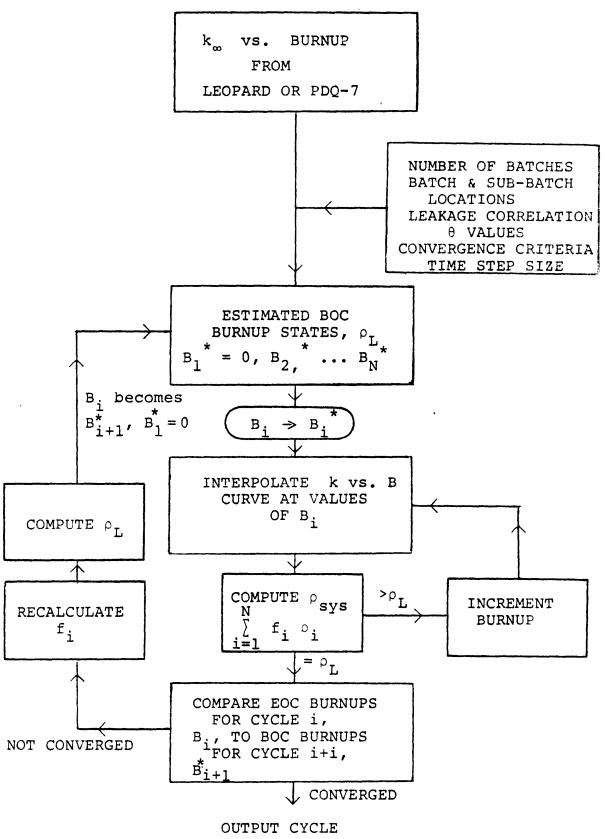
Number of assemblies in a batch and location (peripheral and/or interior) are input parameters

$$\rho_{L} = \alpha + \beta f_{per}$$

$$f_{per} = \sum_{i=1}^{N} f_{i,per}$$

$$f_{i,int} = \frac{\frac{M_{i,int}(k_i)^{\Theta}int}{M_i}}{\sum_{\substack{i=1\\i=1}}^{N} f_i}$$

$$f_{i,per} = \frac{\frac{M_{i,per}}{M_{i}} (k_{i}^{PNL_{i}})^{\theta} per}{\sum_{\substack{i=1 \\ i=1}}^{N} f_{i}}$$





If the k_{∞} versus burnup data input to the code comes from an infinite medium spectrum calculation, the leakage correlation supplied to the code should have an axial leakage term incorporated in the first constant. If the k_{∞} versus burnup data comes from a PDQ-7 bundle calculation, and includes the effect of axial leakage, only radial leakage effects are included in the correlation.

The code was applied to the low-leakage fuel management problem analyzed in Chapter 3. A leakage correlation was developed for application to equilibrium cases by fitting the (near-equilibrium) values of the peripheral assembly powers for both cases available (low-leakage and outin) to their corresponding leakage reactivity values. The k_∞ versus burnup values for 4.30 and 4.44 w/o U-235. Maine Yankee assemblies, generated by LEOPARD, were input for the out-in and low-leakage cases, respectively. The values of θ_{ner} and θ_{int} were both set equal to 2.0, and the non-leakage probability for peripheral batches was taken to be 0.77. A shim penalty of 0.5% $\Delta\rho$ was applied to the low-leakage case to allow for the use of burnable poison in suppressing the radial power peaking associated with this fuel arrangement. The ALARM code was then used to compute the steady-state burnup in both cases; from the burnup the uranium utilization was then estimated. The uranium savings of the low-leakage core relative to the out-in core was found to be 5.574% before adjustment for

the reload enrichment differences. After adjustment, a 2.1% U308 savings resulted. The detailed computer studies reported by CE (M-3) predict a 2.5% saving--a value which includes the effect of having 5.6% less fuel in the lowleakage assemblies, an effect not simulated here. However, the overall assembly-averaged spectrum becomes more thermal under this condition, which tends to compensate for the fuel removal. Thus, our simple first cut at this complex problem yields rather satisfactory agreement especially since no attempt was made at a posteriori fine tuning of any of the model's parameters such as θ and PNL. Α proper analysis would involve the calculation of the $k_{\boldsymbol{\varpi}}$ versus burnup curves for both types of lattices in a full PDQ-7 spatial assembly treatment with the boron rods explicitly included.

4.6 Chapter Summary

In this chapter, simple algorithms have been developed which enable the user to calculate the power splits among batches in a reactor core, and to determine the power in the peripheral zone of the reactor for input into a radial leakage correlation. The algorithms were assembled into a code, ALARM, which can be used to generate batch power histories: i.e., the program fulfills the function of a "poor man's" PDQ. ρ versus burnup curves of arbitrary shape can be input into the code, in which a "curve following" depletion computation is used to fuel management problems will be documented further in Chapter 6.

CHAPTER 5

AXIAL FUEL MANAGEMENT

5.1 Introduction

Previous work at M.I.T. (K-1, F-1) and by many others elsewhere (M-3, C-5, R-2) indicated that improvements in "axial fuel management" could result in uranium savings in PWRs. In this chapter the basic fuel assembly design variables which comprise an axial fuel management strategy will be examined. Several different axial blanket fuel arrangements will be surveyed and the most promising options will be analyzed in detail.

5.2 Analytical Considerations

The goal of improved uranium utilization in PWRs can be achieved by maximizing the reactivity of a fuel bundle over life. Consider a fuel assembly divided into N axial regions (i=1,N); the reactivity of region i can be written as:

$$\rho_{i} = \rho_{0,i} - A_{i} B_{avg} g_{i} / (1/N)$$
 (5.1)

where $\rho_{0,i}$ is the initial reactivity of the fuel in region i, A_i is the slope of the reactivity vs. burnup curve for the fuel in the ith region, B_{avg} is the average burnup of the assembly (MWD/MT), g_i is the cumulative fraction of the total assembly energy generated in the ith region since the start of assembly irradiation, and 1/N is the average cumulative power generated in each region. Thus, the burnup

in region i can be written as:

$$B_{i} = B_{avg} N = \frac{\int_{0}^{t} f_{i}(t) dt}{\int_{0}^{t} \sum_{i=1}^{N} f_{i}(t) dt} = B_{avg} N g_{i} \qquad (5.2)$$

Considering all N regions of the assembly, the set of local reactivity values can be written as:

$$\rho_{1} = \rho_{0,1} - A_{1} B_{avg} N g_{1}$$

$$\rho_{2} = \rho_{0,2} - A_{2} B_{avg} N g_{2}$$

$$\cdot$$

$$\cdot$$

$$\rho_{N} = \rho_{0,N} - A_{N} B_{avg} N g_{N}$$
(5.3)

Applying the equation for power-weighted reactivity; namely,

$$\rho_{\text{sys}} = \begin{bmatrix} \mathbf{N} & \mathbf{f}_{i} & \rho_{i} \end{bmatrix} - \rho_{L}$$
(5.4)

to Eq. 5.3 yields:

$$\rho_{\text{sys}} = \begin{bmatrix} N \\ \sum_{i=1}^{N} f_{i} \rho_{o,i} \end{bmatrix} - \begin{bmatrix} N \\ \sum_{i=1}^{N} A_{i} B_{avg} Ng_{i} f_{i} \end{bmatrix} - \rho_{L}$$
(5.5)

where f_i is the power fraction in region N at the time of evaluation. It is useful to consider the special case of a critical reactor ($\rho_{sys} = 0$) which operated with a time-invariant power profile ($f_i = g_i$) and a uniform fuel loading. For this limiting case:

$$\rho_{\text{sys}} = 0 = \rho_{0} - \left[\sum_{i=1}^{N} AB_{\text{avg}} N f_{i}^{2}\right] - \rho_{L}$$
(5.6)

and the average assembly burnup is given by:

$$B_{avg} = \frac{\rho_{o} - \rho_{L}}{AN\left[\sum_{i=1}^{N} f_{i}^{2}\right]}$$
(5.7)

Clearly, the average burnup can be maximized by minimizing the leakage reactivity, ρ_L , and by establishing as uniform a power profile (all f_i equal) as possible. Since the axial leakage reactivity can be correlated as

$$\rho_{\rm L} = \alpha + \beta f_{\rm per} \tag{5.8}$$

the minimization of ρ_L involves the reduction of power in peripheral regions (i.e., achieving a low value of f_{per} within two or three migration lengths of the ends of the fuel assembly). Minimization of ρ_L conflicts with the minimization of the axial power profile index $\sum_{i=1}^{N} f_i^2$.

The equations become more complicated if more than one type of fuel is present in the assembly. Consider the case where axial regions 1 through M consist of fuel with reactivity ρ_0 and slope A and regions M+1 through N consist of fuel with reactivity ρ'_0 and slope A'. The reactivity equation can be written as:

$$\rho_{1} = \rho_{0,1} - AB_{avg} N f_{1}$$

$$\rho_{2} = \rho_{0,2} - AB_{avg} N f_{2}$$

$$.$$

$$.$$

$$\rho_{M} = \rho_{0,M} - AB_{avg} N f_{M}$$

$$\rho_{M+1} = \rho_{0,M+1} - A'B_{avg} N f_{M+1}$$

$$.$$

$$.$$

$$.$$

$$\rho_{N} = \rho'_{0,N} - A'B_{avg} N f_{N}$$
(5.9)

In this case, the burnup at a given point in time can be written as

$$B_{avg} = \frac{\begin{bmatrix} M & & \\ \vdots = 1 & f_{i} & \rho_{o,i} \end{bmatrix} + \begin{bmatrix} N & & \\ \sum & f_{i} & \rho'_{o,i} \end{bmatrix} - \rho_{L}}{N \begin{bmatrix} A & & \\ \sum & f_{i} & 2 \end{bmatrix} + A' \begin{bmatrix} N & & \\ \sum & f_{i} & 2 \end{bmatrix}}$$
(5.10)

The formulation embodied in Eq. 5.10 provides an approximate way to compare various strategies for fuel management under various assumptions about the power profile. However, there are several salient points which must be addressed regarding multi-zone fueling in the axial direction.

 Substantial changes in the axial power shape typically occur during an assembly's first cycle of residence in the reactor (i.e., 0-10000 MWD/MT). This weakens the assumption that $f_i = g_i$.

2) Lowering the power generated at the ends of the bundle generally involves reducing the enrichment of the fuel on the periphery. Thus, the neutron energy spectrum in these regions, where leakage is also high, may not be sufficiently close to the infinite medium spectrum assumed for the calculation of the ρ vs. burnup curves for these regions; and thermal-hydraulic feedback effects have been neglected.*

3) For end regions consisting of uranium oxide of less-than-natural uranium enrichments (<0.711 w/o U-235) the ρ vs. burnup curves can be highly non-linear.

4) From the thermal-hydraulic standpoint, both the peak and average powers and the detailed axial power shape are important because of the need to satisfy fuel centerline melt and DNBR constraints. Thus, one is not free to arbitrarily pick an axial power shape or fuel loading scheme that optimizes uranium utilization alone.

Because of these considerations, it was decided that only detailed PDQ-7-Harmony depletion analyses of the axial direction should be used for final determination of the uranium savings produced by these interacting effects. Nevertheless, the following general guidelines for axial fuel management strategy were deduced from the analytical model and used to select and screen the cases for the final evaluation:

*Uniform axial temperature distribution assumed

- 1) minimize power in the peripheral core regions
- in non-peripheral regions the power profile should be as flat as possible.

In addition, since the axial leakage can be correlated as

$$\rho_{\rm L} = \alpha + \beta f_{\rm per} \tag{5.11}$$

it may be possible to employ materials (e.g., beryllium, zirconium) that lead to coefficients (α, β) which are smaller than the values typical of assemblies "reflected" by the mixture of stainless steel and water in the upper and lower assembly support structure, and which therefore lead to lower leakage reactivities for the same peripheral power.

5.3 Axial Blanket Results*

Since the uranium utilization calculations are particularly sensitive to the peripheral zone power, investigations have been centered around techniques designed to either reduce the power (and hence ρ_L) or to alter the structural material near the end of the assembly to reduce ρ_L at the same peripheral power. The standard Maine Yankee reactor assembly design described in Appendix A was used for the analysis. The five cases analyzed are shown in Fig. 5.1.

The reference case consisted of a 69 inch (half-core) fueled region containing 3.0 w/o U-235 and an unfueled region which consisted of 28 inches of stainless steel structure and water. The fuel cross-sections were homogenized using

^{*} Uniform axial temperature distribution assumed.

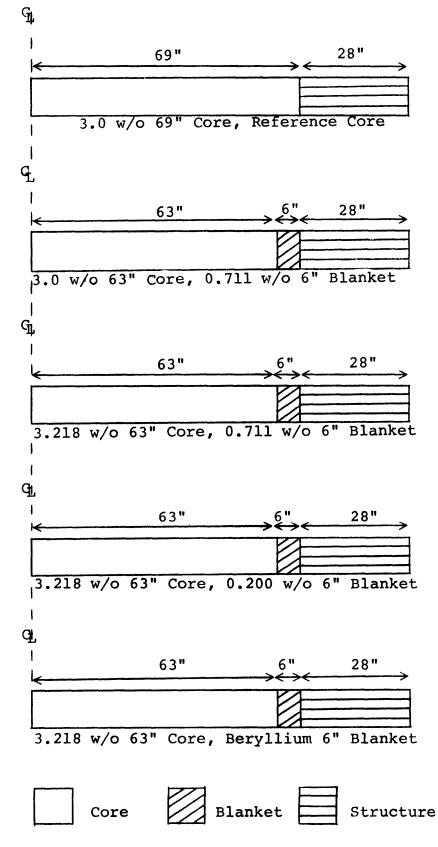


FIG. 5.1 CASES CONSIDERED FOR IMPROVING URANIUM UTIL-IZATION VIA AXIAL FUEL MANAGEMENT

a LEOPARD supercell geometry. For all cases, the total power level of the bundle was kept constant. Table 5.1 shows the values of ρ_{eff} , ρ_{L} , and the axial peak-to-average power ratio in the assembly as a function of time (in hours) at effective full power. The values of reactivity vs. burnup for the reference case were submitted to the ALARM code under conditions of no radial leakage, equal radial power sharing among batches ($\theta_{I} = \theta_{P} = 0.0$), and three-batch fuel management. The results indicate that the spent fuel is discharged after 26,179 hours of irradiation at full power.

Two natural uranium blanket cases were examined. In case 2, 63 inches of 3.0 w/o U-235 occupied the central core region, while the last six inches of the core were replaced with natural uranium (0.711 w/o U-235). Case 3 is essentially the same except that the central core region consists of 3.218 w/o U-235 instead of 3.0 w/o U-235. Thus, Case 3 and the reference case have the same feed-to-product ratio (F/P). The values of ρ_{eff} , ρ_{L} and axial peak-to-average power for Cases 2 and 3 are listed in Tables 5.2 and 5.3 respectively. Both cases have nearly identical leakage reactivities and power profiles over their entire burnup history. The fuel assembly analyzed in case 2 is discharged at 24,860 hours, whereas the fuel assembly analyzed in Case 3 is discharged at 26,797 hours. The relative uranium requirement of these cases will be examined in the section which follows.

Time (Hours at full power)	^ρ eff	ρ _Γ	Peak-to-Average Power Ratio
0	0.24778	0.00314	1.500
125	0.22033	0.00432	1.329
800	0.21048	0.00448	1.318
1500	0.20324	0.00508	1.230
3000	0.18678	0.00651	1.148
4500	0.16951	0.00739	1.128
6000	0.15180	0.00834	1.121
7500	0.13399	0.00902	1.107
9000	0.11575	0.00983	1.106
10500	0.09743	0.01037	1.091
12000	0.07865	0.01117	1.098
13500	0.05949	0.01217	1.087
15000	0.04005	0.01236	1.087
16500	0.01969	0.01316	1.094
18000	-0.00101	0.01365	1.083
19500	-0.02242	0.01447	1.093
21000	-0.04477	0.01514	1.090
22500	-0.06746	0.01570	1.081
24000	-0.09091	0.01643	1.083
25500	-0.11473	0.01689	1.072
27000	-0.13851	0.01735	1.064
28500	-0.16021	0.01669	1.056
30000	-0.18159	0.01724	1.031

,

TABLE 5.1 REFERENCE CASE BURNUP RESULTS

Time (Hours at full power)	ρ_{eff}	^ρ L	Peak-to-Average Power Ratio
0	0.24738	0.00147	1.590
125	0.21941	0.00197	1.431
800	0.20916	0.00209	1.402
1500	0.20141	0.00239	1.313
3000	0.18378	0.00315	1.215
4500	0.16536	0.00370	1.193
6000	0.14666	0.00426	1.182
7500	0.12777	0.00477	1.172
9000	0.10859	0.00524	1.161
10500	0.08930	0.00573	1.149
12000	0.06939	0.00629	1.155
13500	0.04926	0.00672	1.143
15000	0.02848	0.00733	1.151
16500	0.00709	0.00784	1.144
18000	-0.01469	0.00837	1.141
19500	-0.03740	0.00914	1.157
21000	-0.06105	0.00962	1.139
22500	-0.08501	0.01033	1.148
24000	-0.10949	0.01094	1.141
25500	-0.13401	0.01152	1.131
27000	-0.15663	0.01151	1.079
28500	-0.17850	0.01198	1.072
30000	-0.19992	0.01232	1.064

TABLE 5.2	NATURAL	URANIUM	BLANKETED	ASSEMBLY	BURNUP	RESULTS,
	3.0 W/O	CORE REC	GION			

Time (Hours at full power)	^ρ eff	$\frac{\rho_{L}}{2}$	Peak-to-Average Power Ratio
0	0.25574	0.00143	1.591
125	0.22836	0.00193	1.426
800	0.21819	0.00205	1.402
1500	0.21076	0.00732	1.319
3000	0.19401	0.00303	1.2 16
4500	0.17649	0.00355	1.194
6000	0.15868	0.00408	1.183
7500	0.14069	0.00456	1.173
9000	0.12241	0.00503	1.164
10500	0.10400	0.00546	1.155
12000	0.08507	0.00597	1.156
13500	0.06592	0.00639	1.145
15000	0.04609	0.00695	1.154
16500	0.02564	0.00743	1.146
18000	0.00475	0.00795	1.146
19500	-0.01732	0.00865	1.160
20000	-0.03984	0.00912	1.144
22500	-0.06304	0.00980	1.153
24000	-0.08686	0.01037	1.147
25500	-0.11087	0.01095	1.140
27000	-0.13317	0.01099	1.086
28500	-0.15510	0.01146	1.084
30000	-0.17655	0.01179	1.070

TABLE 5.3 NATURAL URANIUM BLANKETED ASSEMBLY BURNUP RESULTS 3.218 W/O CORE REGION

In Case 4, the natural uranium blanket of Case 3 is replaced with a depleted uranium (0.2 W/O U-235) blanket. The reactivity history for this case is shown in Table 5.4. The peak to average power ratios are higher for this case than for Case 3 bacause the power in the last six inches is somewhat depressed. However, the leakage reactivity, ρ_L , is also reduced significantly, and this implies more efficient neutron utilization. The discharge time for Case 4 is computed by the ALARM code to be 26466 hours.

In the fifth and final case, the depleted uranium blanket fuel pellets of Case 4 are replaced by beryllium metal slugs. The cross sections for this region were generated using the HAMMER code since the available LEOPARD library did not include beryllium.

The beryllium region has a higher fast neutron albedo than stainless steel-containing regions, and as a result, more neutrons are returned to the core region. However, this causes the power to rise in the peripheral regions, which forms an offsetting increase in neutron leakage. The reactivities and peak-to-average power ratios for the beryllium blanket case are shown in Table 5.5.

The leakage reactivities for the beryllium blanket case are comparable to those of the reference case over life and even drop lower than those of the reference case near the end of life. Additional benefit comes from the remarkably flat power profile in the fueled region over life. A lower

Time (Hours at full power)	^p eff	ρ _L	Peak-to-Average Power Ratio
0	0.25565	0.00091	1.612
125	0.22815	0.00122	1.450
800	0.21788	0.00132	1.424
1500	0.21033	0.00151	1.339
3000	0.19329	0.00202	1.234
4500	0.17548	0.00243	1.212
6000	0.15743	0.00286	1.199
7500	0.13916	0.00327	1.190
9000	0.12065	0.00368	1.180
10500	0.10198	0.00409	1.172
12000	0.08281	0.00454	1.171
13500	0.06342	0.00495	1.160
15000	0.04328	0.00549	1.172
16500	0.02259	0.00594	1.160
18000	0.00013	0.00649	1.166
19500	-0.0297	0.00711	1.175
21000	-0.04370	0.00764	1.163
22500	-0.06721	0.00831	1.164
24000	-0.09121	0.00889	1.162
25500	-0.11517	0.00948	1.153
27000	-0.13736	0.00956	1.095
28 5 00	-0.15934	0.01018	1.099
30000	-0.18068	0.01055	1.081

TABLE 5.4 DEPLETED URANIUM BLANKETED ASSEMBLY BURNUP RESULTS 3.218 W/O CORE REGION

3.218 W/O	CORE REGION		
Time (Hours at Full Pow er)	ρ e ff		Peak-to-Average Power Ratio*
0	0.25583	0.00414	1.442
125	0.22849	0.00542	1.294
800	0.21832	0.00566	1.276
1500	0.21091	0.00632	1.201
3000	0.19414	0.00792	1.112
4500	0.17647	0.00889	1.095
6000	0.15842	0.00978	1.087
7500	0.14009	0.01048	1.090
9000	0.12138	0.01108	1.122
10500	0.10246	0.01158	1.139
12000	0.08292	0.01216	1.158
13500	0.06307	0.01251	1.152
15000	0.04233	0.01313	1.167
16500	0.02093	0.01346	1.150
18000	-0.01237	0.01393	1.144
19500	-0.02459	0.01441	1.135
21000	-0.04850	0.01463	1.105
22800	-0.07347	0.01506	1.090
24000	-0.09893	0.01519	1.053
25500	-0.12381	0.01502	1.032
27000	-0.14733	0.01476	1.047
28500	-0.17064	0.01486	1.040
30000	-0.19300	0.01471	1.049

5.5 BERYLLIUM METAL BLANKETED ASSEMBLY BURNUP RESULTS, 3.218 W/O CORE REGION

* Since the fuel pellet is shorter, multiply results by 69/63 to compare active linear power to other cases.

peak-to-average power ratio permits better fuel utilization near the ends of the fueled region. However, the fuel stack length is shorter, and hence the peak-to-average power value must be multiplied by 69/63 to obtain linear power values which can be compared to the reference case on an absolute basis. The ALARM code calculates an irradiation time of 26096 hours for the beryllium-blanketed case. Fig. 5.2 shows the leakage reactivities for four cases as a function of time.

5.4 Uranium Utilization for Assemblies with Axial Blankets

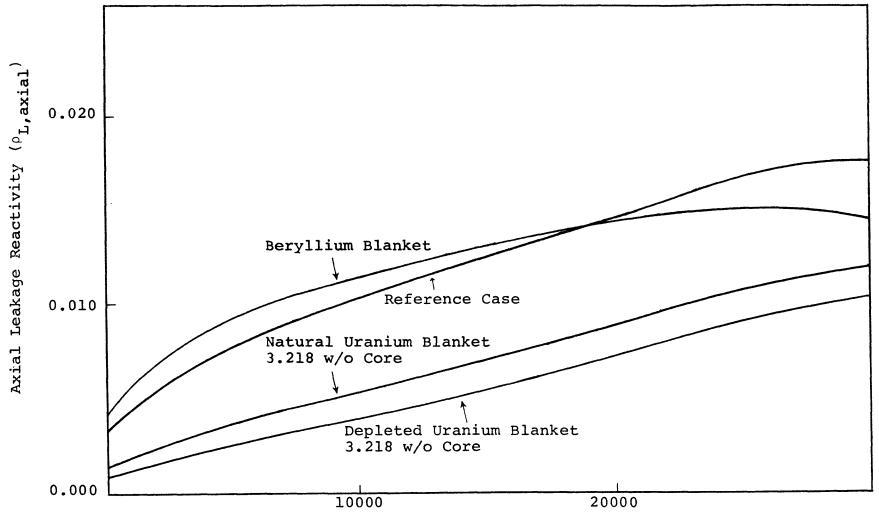
The feed-to-product ratio for the cases analyzed in Section 5.3 can be written as

$$\frac{F}{P} = \frac{V_{\text{core}}}{V_{\text{total}}} \left\{ \frac{\varepsilon_{\text{core}} - 0.2}{0.711 - 0.2} \right\} + \frac{V_{\text{blanket}}}{V_{\text{total}}} \left\{ \frac{\varepsilon_{\text{blanket}} - 0.2}{0.711 - 0.2} \right\}$$
(5.12)

where V_{core} is the volume of the core region, $V_{blanket}$ is the volume of the blanket region, V_{total} is the sum of V_{core} and $V_{blanket}$, and ε_{core} and $\varepsilon_{blanket}$ are the core and blanket enrichments, respectively. The feed-to-product ratios for the five cases and the uranium (natural) usage are shown in Table 5.6.

Table 5.7 shows the relative hours at effective full power, the relative natural uranium feed and the relative uranium usage $\left(\frac{\text{MTF}}{\text{MWD}}\right)$ for each case.

The natural uranium blanket case with the 3.0 w/c core falls short of the reference case discharge time by about



Time (Hours) at Effective Full Power

FIG. 5.2 AXIAL LEAKAGE REACTIVITY (p FOR REFERENCE AND BLANKET (Axial) VS. TIME AT EFFECTIVE FULL POWER CASES

TABLE 5.6 FEED-TO-PRODUCT RATIOS AND RELATIVE NATURAL URANIUM USAGE FOR AXIAL FUEL MANAGEMENT CASES

Case	Type	<u>F/P</u>	Relative Natural Uranium Usage
l	3.0 W/O 69" Core Reference Core	5.47945	1.00000
2	3.0 W/O 63" Core 0.711 W/O 6" Blanket	5.08993	0.92891
3	3.218 W/O 63" Core 0.711 W/O 6" Blanket	5.47945	1.00000
4	3.218 W/O 63" Core 0.200 W/O 6" Blanket	5.39246	0.98413
5	3.218 W/O 63" Core Beryllium 6" Blanket	5.39246	0.98413

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Case	Туре	Relative Hours at Effective Full Power	Relative Natural Uranium Feed	Relative Uranium Requirements per Unit Energy Output
1	3.0 W/O 69" Core Reference Core	1.00000	1.00000	1.00000
2	3.0 W/O 63" Core 0.711 W/O 6" Blanke	0.94962 et	0.92891	0.97819
3	3.218 W/O 63" Core 0.711 W/O 6" Blanke		1.00000	0.97693
4	3.218 W/O 63" Core 0.200 W/O 6" Core	1.01096	0.98413	0.97346
5	3.218 W/O 63" Core Beryllium 6" Blanke	0.99682 et	0.98413	0.98727

TABLE 5.7 RELATIVE URANIUM REQUIREMENTS FOR NORMALIZED POWER OUTPUT FOR VARIOUS BLANKET STRATEGIES

5.1%, but produces a uranium Baving of 2.19%. The natural uranium blanket case with the 3.218 w/o core surpasses the reference case discharge time by 2.36% and produces a uranium saving of 2.31%. In order to match the cycle length of the reference case, the enrichment of the core should be adjusted to a value between 3.0 and 3.218 W/O U-235. For this case the uranium saving would lie between 2.19% and 2.31%.

The Case 4 configuration with the depleted uranium blanket extends the cycle length by about 1.01%, and thus would require a lower enrichment than the 3.218 w/o value used in the core zone. This would reduce the uranium saving from the 2.65% value quoted by perhaps as much as 0.05%. However, to first order, these effects are negligible.

Case 5 demonstrates that a 1.13% uranium saving is possible using a beryllium blanket. The saving results from the decrease in leakage reactivity attributable to the higher albedo of beryllium relative to stainless steel plus water. Additionally, the beryllium-blanketed core achieves flatter power profiles over life, which means that the fuel near the end of the core is more fully utilized. This decreases the axial profile index of Eq. 5.6.

Table 5.8 shows the maximum/average power ratio at BOL for the blanketed cases relative to the reference case. Clearly, all of the blanketed cases have higher axial peaking factors than the reference case at BOL.

TABLE 5.8	MAXIMUM/AVERAGE	POWER AT BOL FOR BLANKETED	CASES
		REFERENCE CASE	

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Case	Туре	Relative Maximum/ Average Power*
1	3.0 W/O 69" Core Reference Core	1.000
2	3.0 W/O 63" Core 0.711 W/O 6" Blanket	1.060
3	3.218 W/O 63" Core 0.711 W/O 6" Blanket	1.061
4	3.218 W/O 63" Core 0.200 W/O 6" Blanket	1.075
5	3.218 W/O 63" Core Beryllium 6" Blanket	1.053

* Maximum/Avg power divided by Maximum/Avg power for reference core.

This limitation at BOL can be compensated for by one or a combination of several strategies:

- enrichment zoning of the core region to redistribute power
- 2) inclusion of burnable poison (e.g., gadolinium) in the oxide fuel in the regions where power peaks would otherwise occur
- Use of annular fuel pellets over part of the core length to adjust the local fuel-to-moderator ratio

5.5 Chapter Summary

Improvement of uranium utilization in PWRs by use of axial fuel arrangement schemes ideally involves a concurrent reduction of axial leakage and a flattening of power in fueled zones to efficiently burn all segments of the fuel. The use of natural uranium blankets six inches in length can result in uranium savings of about 2.3%, whereas the use of depleted uranium blankets of the same length can result in ore savings of up to 2.7%.

Replacing the last six inches of fuel with beryllium metal slugs can yield a uranium saving of about 1.1%. These results should eventually be subjected to a more careful analysis using transport theory methods since diffusion theory (particularly in only two groups) may not be adequate to compute leakage near the material interfaces at the ends of the assembly. In addition, mitigation of the power peaking which accompanies the use of blankets (by enrichment zoning, the use of zoned burnable poison, or zones of annular fuel pellets) should be examined since it is not clear that an optimum configuration has been reached at this point.

CHAPTER 6

RADIAL FUEL MANAGEMENT

6.1 Introduction

In this chapter, the advanced linear reactivity model is applied to problems relating to uranium utilization in extended cycle operation (i.e., l.5 year cycles vs. l.0 year cycles). The ALARM code is used to analyze the Maine Yankee reactor for extended cycle and low-leakage extended cycle operation relative to current operating conditions. Finally, the use of assemblies of natural uranium fuel on the core periphery is examined to determine if uranium savings are possible using this strategy.

6.2 Extended Cycle Length/Burnup

In recent years, utilities have been placing more emphasis on extended cycle lengths to improve overall power plant economics (S-3, S-8, F-2). However, the use of extended cycles in place of normal cycles (e.g., 18 month cycles vs. 12 months cycles) can cause a decrease in uranium utilization depending on the discharge burnup and the number of staggered fuel batches.

The discharge burnup of the fuel can be written as:

$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A\left\{\sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j}\right\}}$$
(6.1)

where $\rho_{\rm L}$ is the leakage reactivity and N is the number of in core batches (constrained at this point to integer values). If all assemblies in the reactor share power equally $(f_i = \frac{1}{N})$, the cycle schedule index in Eq. 6.1 is given by

$$CSI = \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} = \frac{N+1}{2N}$$
(6.2)

In this case, $N = 1, 2, \ldots, \infty$.

For an integer number of batches, the cycle burnup is simply the discharge burnup divided by the number of batches. A utility operating with a fixed discharge burnup constraint may wish to vary the cycle burnup (and hence the time between refuelings). For a fixed enrichment this can only be accomplished by changing the reload fraction of the core. This will result in a non-integer number of batches in the core and ultimately it will affect the uranium utilization.

For a fixed reload enrichment, the number of reload assemblies, A_R , is related to the total number of assemblies in the core, A_m , by

$$A_{R} = A_{T} \quad \frac{B_{CYC}}{B_{dis}}$$
(6.3)

where B_{cyc} is the average burnup of the fuel in one cycle and B_{dis} is the average discharge burnup of the fuel. If the discharge burnup is fixed, the cycle burnup can be adjusted by changing the number of reload assemblies. The cycle schedule index can be derived for any combination of batches in a core (see Appendix F). If the power sharing among the assemblies is optimal, the cycle schedule index reduces to:

$$CSI \cong \frac{N+1}{2N}$$
(6.4)

where N is now an integer or noninteger number given by:

$$N = \frac{A_{\rm T}}{A_{\rm R}} \tag{6.5}$$

Applying Eq. 6.3 gives the discharge burnup as:

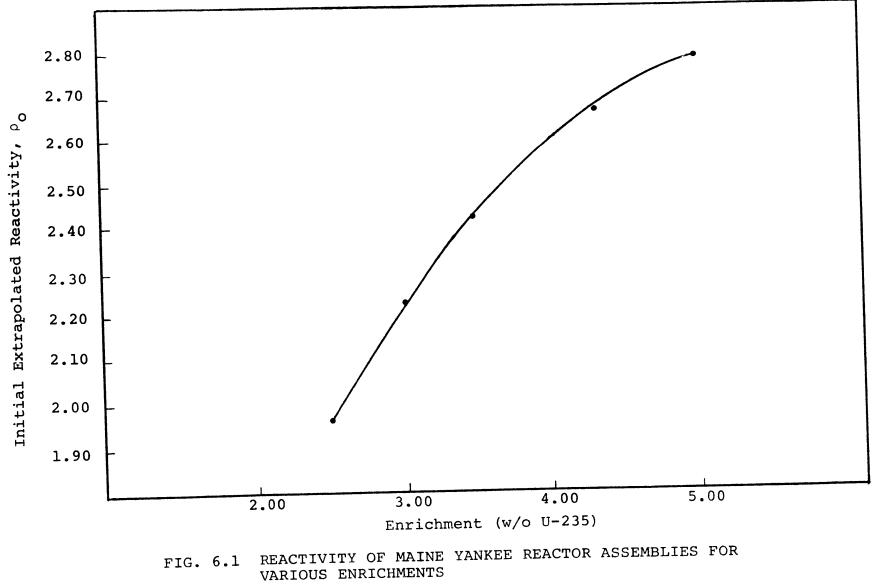
$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A \left[\frac{B_{dis}}{B_{cyc}} + 1\right]}$$
(6.6)
$$2 \left[\frac{B_{dis}}{B_{cyc}}\right]$$

Maine Yankee reactor assembly burnup calculations have been analyzed using LEOPARD supercell geometry for enrichments ranging from 2.50 w/o U-235 to 5.00 w/o U-235. In this range, the initial extrapolated reactivity can be correlated as:

$$\rho_0 = 0.0103556 + (0.0960171)\epsilon - (0.0084899)\epsilon^2$$
 (6.7)

and the slope of the reactivity vs. burnup curve, A, can be correlated as

$$A = 1.5617773 \times 10^{-5} - (0.2757294 \times 10^{-5})\varepsilon + (0.0184100 \times 10^{-5})\varepsilon^2 (\frac{MWD}{MT})^{-1}$$
(6.8)



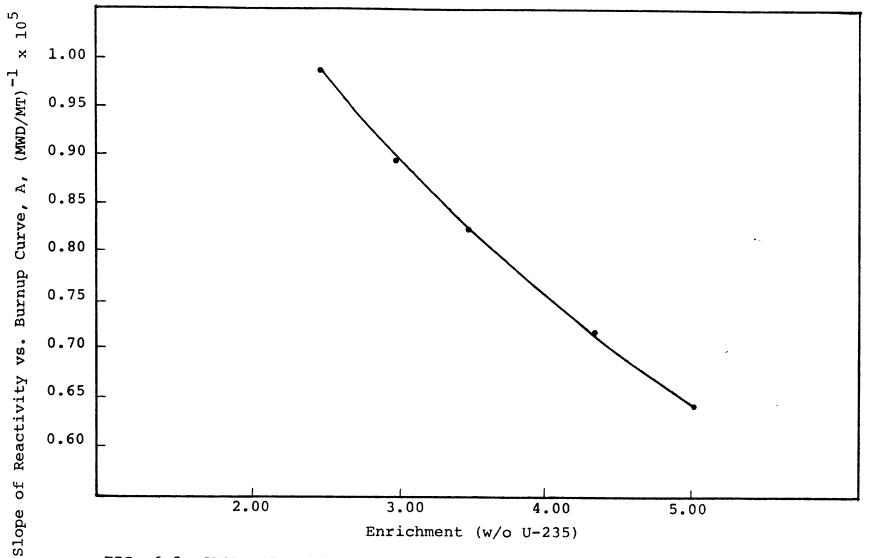


FIG. 6.2 SLOPE OF REACTIVITY VS. BURNUP CURVE FOR MAINE YANKEE REACTOR ASSEMBLIES FOR VARIOUS ENRICHMENTS

where ε is the enrichment of the feed in w/o U-235. The curves are plotted in Figs. 6.1 and 6.2.

Thus, Eq. 6.6 becomes:

$$\frac{B_{dis} + B_{cyc}}{2} = \frac{0.0103556 + (0.0960171)\varepsilon - (0.0084899)\varepsilon^2 - \rho_L}{\{1.561773 - (0.2757294)\varepsilon + (0.0184100)\varepsilon^2 \times 10^{-5} \}}$$
(6.9)

The uranium utilization in MWD per Metric Ton of Feed $(\frac{MWD}{MTF})$ is given by

$$U = \frac{B_{dis} \left(\frac{MWD}{MTP}\right)}{\left(\frac{F}{P}\right)} = \frac{0.511 B_{dis}}{(\epsilon - 0.2)}$$
(6.10)

for an 0.2 w/o U-235 tails assay.

The cycle burnup can be related to the calendar time between startups by:

$$B_{cyc} = \frac{(P_{TH})(T-T_R)(CF) 365}{(A_T)(M)}$$
(6.11)

where:

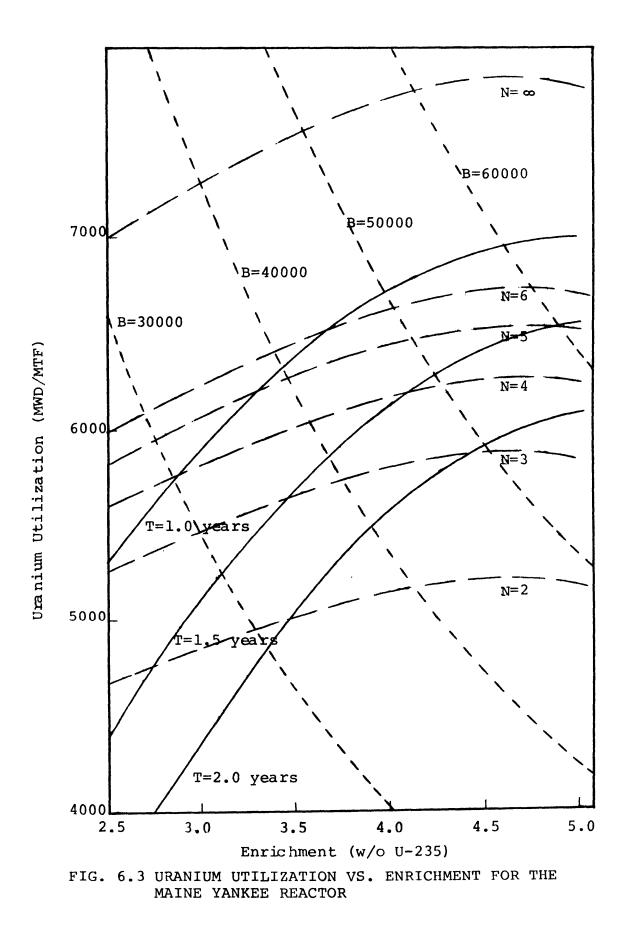
$$P_{TH}$$
 = Reactor thermal power rating (Mwth)
T = Time between startups (years)
 T_R = Refueling downtime (years)
 $T-T_R$ = Time the reactor is available to produce power
 CF = Availability-based capacity factor
 A_T = Total number of assemblies in reactor
M = Heavy metal loading per assembly (Metric Tons
Uranium)

Eq. 6.7-6.11 were combined to estimate the uranium utilization of the Maine Yankee reactor for a variety of enrichments, discharge burnups, cycle times and batch numbers. The values for the reactor parameters of Eq. 6.13 are:

 $P_{th} = 2630 \text{ Mwth}$ $T_R = 0.115068 \text{ years}$ $C_F = 0.75$ $A_T = 217$ M = 0.388 MT

In addition, a leakage reactivity of 0.042 was assumed and a tails assay of 0.2 w/o U-235 was used.

The results are shown in Fig. 6.3. The Maine Yankee reactor is currently entering cycle 4 of operation with a capacity factor of about 0.65. Nevertheless, if a current operating point could be defined for this figure it would This would correspond to a cycle be at $\varepsilon = 3.04$ and N = 3. time (startup to startup) of ~1.3 years, and a uranium utilization of 5500 MWD/MTF. Clearly, the graph demonstrates that substantial improvements in uranium utilization with cycle times acceptable to utilities (T > 1 year) can only be achieved by increasing the discharge burnup and the number of in-core batches. For example, a 5 batch core with a 1.3 year cycle would have a discharge burnup of about 53000 MWD/MTP and a uranium utilization of 6500 MWD/MTF. This represents an 18.2% improvement in uranium utilization, or



a 15.4% reduction in uranium requirements relative to current operating conditions.

6.3 Low-Leakage Extended Cycle Fuel Management

The ALARM code has been used to analyze the effects of extended cycle length and low-leakage fuel management on the uranium utilization of the Maine Yankee reactor. Three cases of practical interest were examined. The first case (the reference case) consisted of a reload of 72 assemblies of 3.0 w/o U-235 fuel operated in a normal out-in refueling In the second and third cases, 72 assemblies of 3.4 mode. w/o U-235 were used as the reload for out-in and low-leakage fuel management schemes. The radial leakage correlation for the Maine Yankee reactor developed in Chapter 4 was used, and a constant axial leakage reactivity allowance of 0.012 was included. The cycle burnups, discharge burnups, and uranium utilizations for the three cases are listed in Table In the first two cases, the 44 peripheral assembly 6.1. locations were loaded with fresh fuel, whereas in the lowleakage case, third cycle fuel was used for the peripheral locations. A deficit for burnable poison residual was not included in the extended cycle cases. The relative uranium requirements $\left(\frac{MTF}{MWD}\right)$ for the three cases are listed in Table 6.2.

TABLE 6.1	COMPARISON OF	EXTENDED CYCLE	E AND LOW-LEAF	KAGE EXTENDED	CYCLE FU	EL MANAGEMENT
	WITH CURRENT,	THREE-BATCH, C	OUT-IN FUEL MA	ANAGEMENT		

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Case	ε (W/O U-235)	Cycle Burnup (MWD/MTP)	Cycle Length (Years)	Discharge Burnup (MWD/MTP)	U (MWD/MTP)
Current Out-In	3.00	10304	1.32	30632	5590
1.5 year cycle Out-In	3.40	12028	1.522	35759	5710
l.5 year cycle Low-Leakage	3.40	12188	1.540	36734	5866

TABLE 6.2RELATIVE URANIUM REQUIREMENTS FOR EXTENDED CYCLE,
LOW-LEAKAGE EXTENDED CYCLE, AND REFERENCE CASE

Case	Uranium Requirement (MTF/MWD)	Relative Uranium <u>Requirement</u>	
Reference 3.0 W/O Out-In	1.789×10^{-4}	1.000	
Extended Cycle 3.4 W/O Out-In	1.751×10^{-4}	0.979	
Extended Cycle 3.4 W/O Low-Leakage	1.705×10^{-4}	0.953	

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The use of extended cycle fuel management (t_{cycle} - 1.5 years) results in a 2.1% reduction in uranium requirements. Since extended cycle designs generally require more burnable poison than one year cycle designs, which implies larger end-of-cycle residuals, this gain must be considered optimistic. If the extended cycle design is operated in a low-leakage configuration a reduction of 4.7% in uranium requirements results, again neglecting the differential in burnable poison residual. Thus, the low-leakage configuration offers a possible 2.5% reduction in uranium requirements for extended cycle designs. However, the low-leakage extended cycle design may have larger poison residuals than the out-in scatter extended cycle design, thereby reducing the gains.

6.4 Natural Uranium Blankets

Investigations have been performed to determine the effectiveness of natural uranium blankets in improving the uranium utilization of pressurized water reactors. The reference design is the CE-System 80 reactor with five fuel batches and an initial enrichment of 4.34 W/O U-235. The reactivity equations for the reference design can be written as:

$$\rho_{1} = \rho_{0} - AB_{C}N f_{1}$$

$$\rho_{2} = \rho_{0} - AB_{C}N(f_{1} + f_{2})$$

$$\rho_{3} = \rho_{0} - AB_{C}N(f_{1} + f_{2} + f_{3})$$

$$\rho_{4} = \rho_{0} - AB_{C}N(f_{1} + f_{2} + f_{3} + f_{4})$$

$$\rho_5 = \rho_0 - AB_C N(f_1 + f_2 + f_3 + f_4 + f_5)$$
(6.12)

The end-of-cycle reactivity condition is given by:

$$\sum_{i=1}^{5} f_{i} \rho_{i} = \rho_{L} , \qquad (6.13)$$

where $\rho_{\rm L}$ is the radial leakage reactivity, which has been correlated for the system 80 design from published data (CEND-380) to be:

$$\rho_{\rm L} = 0.01126 + 0.2214 \, \rm f_{per}$$
 , (6.14)

where f_{per} is the fraction of the total core power generated in the peripheral assembly locations. The equation for cycle burnup becomes:

$$B_{C} = \frac{\rho_{O} - 0.01126 - 0.2214 \text{ f}_{per}}{5A \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} \right\}}$$
(6.15)

If one-fifth of the core is replaced every cycle the feed to product ratio is given by:

$$\left(\frac{F}{P}\right) = \frac{\varepsilon - 0.2}{0.771 - 0.2}$$
 (6.16)

and the cycle burnup in MWD/MTF is given by:

$$B_{C} \left(\frac{MWD}{MTF}\right) = \frac{\rho_{o} - 0.01126 - 0.2214 \text{ f}_{per}}{5A \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} \right\} \left\{ \frac{4.34 - 0.2}{0.511} \right\} \frac{1}{5}}$$
(6.17)

The reactivity equations for the natural uranium blanket case can be written as:

$$\rho_{1} = \rho_{0} - AB_{C}N f_{1}$$

$$\rho_{2} = \rho_{0} - AB_{C}N(f_{1} + f_{2})$$

$$\rho_{3} = \rho_{0} - AB_{C}N(f_{1} + f_{2} + f_{3})$$

$$\rho_{4} = \rho_{0} - AB_{C}N(f_{1} + f_{2} + f_{3} + f_{4})$$

$$\rho_{5} = \rho_{0,unat} - A' B_{C}N f_{5}$$
(6.18)

When the EOC reactivity balance (Eq. 6.2) is applied, the cycle burnup is given by:

$$B_{C} = \frac{(1 - f_{5})\rho_{0} + f_{5}\rho_{0,unat} - 0.01126 - 0.2214 f_{5}}{5A \left\{ \sum_{i=1}^{2} \sum_{j=i}^{2} f_{i}f_{j} \right\} + 5A' \left\{ f_{5}^{2} \right\}}$$
(6.19)

If one-fifth of the core is replaced with enriched fuel each cycle and the natural uranium zone is replaced every M cycles, the cycle burnup in MWD/MTF can be written as:

$$B_{C}(\frac{MWD}{MTF}) = \frac{(1-f_{5})\rho_{0} - f_{5}\rho_{0,unat} - 0.01126 - 0.2214 f_{5}}{\left[5A\left\{\sum_{i=1}^{\ell}\sum_{j=i}^{\ell}f_{i}f_{j}\right\} + 5A\left\{f_{5}^{2}\right\}\left[\left\{\frac{\varepsilon - 0.2}{0.711 - 0.2}\right\}\frac{1}{5} + \frac{1}{5M}\right]}$$
(6.20)

The ratio of Eq. 6.20 and Eq. 6.17 gives the relative uranium feed use for the blanketed core with respect to the normal 5-cycle equilibrium core. In the normal five batch core, the peripheral assemblies are assumed to be fifth cycle fuel bundles and the remaining four batches share power equally. In the natural uranium blanket case, the four interior core batches also share power equally. Typical parameters for the lattices are: $\rho_{0} = 0.26614$ $A = 0.7154 \times 10^{-5} (MWD/MT)^{-1}$ $\rho_{0,unat} = -0.11934$ $A' = 0.48405 \times 10^{-5} (MWD/MT)^{-1}$

The use of $\rho_{o,unat} = -0.11934$ throughout all cycles of the reactor is clearly optimistic, and is meant to give the natural uranium blanket case the maximum benefit of the doubt. In addition, the natural uranium blanket is optimistically assumed to have an in-core residence time of 10 years (M = 10) to minimize the feed requirements for the blanket case. The ratio of Eq. 6.9 to Eq. 6.6 is plotted as a function of the peripheral zone power fraction in Fig. 6.4.

Even under the best conditions, the natural uranium blanket case requires about 1% more ore than when an equivalent "spent fuel" blanket is used. Representative peripheral zone power fractions for fuel of this type range between 0.10 and 0.15. The advantage of the "spent fuel" over the UNAT blanket under these conditions ranges from 2% to 6%.

If power fractions are to be achieved in natural uranium blankets which are substantially lower than those that can be achieved by most representative 5-batch fuel arrangements, the attendant mid-core power peaking would probably lead to the use of more shim material, leaving residuals

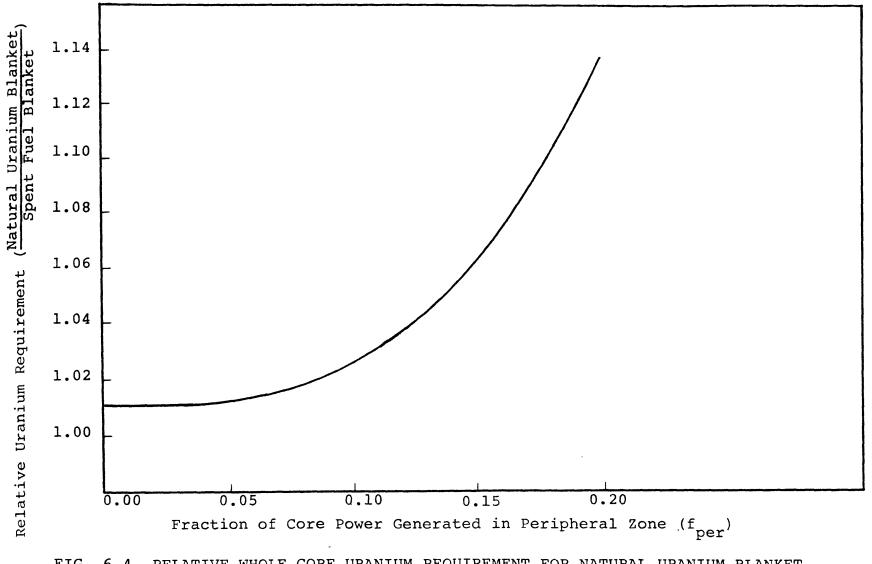


FIG. 6.4 RELATIVE WHOLE CORE URANIUM REQUIREMENT FOR NATURAL URANIUM BLANKET CASE RELATIVE TO SPENT FUEL BLANKET CASE

which would override any possible blanket-related advantage. The conclusion is that for a fixed core size, natural uranium blankets cannot compete with "spent fuel" blankets, i.e., an N-batch core with a natural uranium blanket cannot compete with an (N + 1) batch core which uses its oldest batch as a "blanket".

6.5 Chapter Summary

The uranium utilization for extended cycle fuel management schemes depends on the choice of discharge burnup, cycle burnup (hence number of fuel batches), reload enrichment and leakage reactivity. Only two of these first three parameters can be independently specified; usually the discharge burnup and the cycle burnup (and therefore the cycle length). The leakage reactivity can then be changed by decreasing the amount of fresh fuel on the core periphery.

The calculations of Fig. 6.3 show that the use of extended cycles will increase the uranium utilization $(\frac{MWD}{MTF})$ if the batch number, N, remains constant and the discharge burnup is increased. However, if the discharge burnup remains constant, the use of extended cycles will result in a decrease in uranium utilization and hence an increase in uranium requirements.

The ALARM code shows that the use of extended cycle (1.5 years vs. 1.32 years) out-in/scatter fuel management in the Maine Yankee Reactor will result in a reduction in

uranium requirements of 2.1% if the burnup is increased to approximately 36000 MWD/MT. Concurrent use of low-leakage fuel management in the extended cycle design, however, will result in a reduction of uranium requirements of 4.7% relative to the base case. Overall utility system economics may dictate the use of extended cycle operation. Assumptions about capacity factors and refueling downtime are necessary to compare cycle time and cycle burnup. For extended burnup core designs, the low-leakage scheme requires 2.6% less uranium than the out-in/scatter extended cycle design. If the burnup constraint can be increased to about 53000 MWD/MT, with a concurrent shift to 5 batch fuel management, a reduction of 15.4% in uranium requirements may be possible. Total uranium requirements for high burnup, low-leakage fuel management may (optimistically) approach values 18% lower than under current PWR operating conditions. An analysis which includes burnable poisons in the fuel to reduce peaking factors is necessary to determine burnable poison residuals, which may be high. Also, at higher burnups (~50000 MWD/MT) $\Delta\rho_{T_{\rm c}}$ ($\rho_{T_{\rm c}}({\rm out-in})$ - $\rho_{\rm L}\,({\rm low-leakage})) \; {\rm is} \; {\rm a} \; {\rm smaller} \; {\rm fraction} \; {\rm of} \; \rho_{\rm O} \; , \; {\rm since} \; \rho_{\rm O} \; {\rm is}$ higher for high enrichments. Thus, the advantages of a low-leakage design may be reduced.

Replacing peripheral assemblies with natural uranium blankets does not produce uranium savings even under very optimistic conditions. An N-batch core with a natural uranium blanket has a higher feed requirement than an N + 1

batch core with the oldest batch located on the core peri-

The Hatch-2 (BWR) design calls for a peripheral row of natural uranium assemblies for the initial core, but switches to older fuel in later cycles. The discharged natural uranium bundles can be used in the first core of other BWRS (C-6). However, it does appear that the natural uranium assemblies cannot compete with spent fuel assemblies. The BWR bundles are only 6" in width, and this implies, but does not prove, that using split assemblies in a PWR, in which only the outer quadrants are natural uranium, would be at a similar disadvantage. Even so, the use of smaller assemblies in PWRs (1/4 the current assembly size) may enable lower peaking factors to be achieved in the core, and better low-leakage fuel management schemes to be developed--schemes that offer lower uranium requirements.

CHAPTER SEVEN

FUEL-TO-MODERATOR RATIO EFFECTS

7.1 Introduction

In this chapter, the effects of spectrum changes brought about by changes in the fuel-to-moderator ratio of the fuel lattice will be examined. The fuel-to-moderator ratio of the Maine Yankee Reactor is optimized with regard to uranium utilization by varying the lattice pitch, and at the optimized pitch, the effects of density changes in the fuel are analyzed. A brief discussion of resonance integrals and annular fuel is included to help clarify some recent contradictory results. Finally, the equations which describe "mechanical spectral shift" techniques are derived, and several cases of theoretical and practical interest are analyzed. By "mechanical spectral shift" we mean any process used to change lattice pitch and/or to displace coolant and thereby vary the fuel-tomoderator ratio. Detailed consideration is not given to the actual means used to implement spectral shift, except that the concepts evaluated correspond to pin pulling and bundle reconstitution. There have been as yet no suggestions of a practical scheme for accomplishing continuous mechanical spectral shift. This option was considered interesting, however, as a limiting hypothetical case. Moreover, the neutronic equivalence of pin pulling with bundle reconstitution and discontinuous mechanical shift (from the standpoint of uranium utilization) is demonstrated in Appendix C.

Similarly, we have not assessed the thermal/hydraulic consequences of these evaluations; in some cases the changes examined are so large as to lead one to expect serious difficulties. However, we are again interested in defining limiting cases.

7.2 Optimization of $V_{f/m}$ for Fixed Lattice Designs

In any reactor design, there are several parameters which can influence the fuel-to-moderator ratio of the core; namely, fuel rod pitch, pellet diameter, fuel density, and the presence of voids in the pellet (annular pellets). The goal is to find the best H/U atom ratio for improving uranium utilization without exceeding the limitations imposed by the moderator temperature coefficient. Calculations by several investigators (B-1, M-1, R-1, M-5) indicate that optimizing the fuel-to-moderator ratio can improve uranium utilization by 1-5%. However, the results are reactor-specific--they depend on the current fuel-to-moderator ratio of the design analyzed. Some reactors may already be operating at or near the optimum conditions, while other lattices may leave considerable margin for improvement. Thus, case-specific uranium savings cannot necessarily be applied across the board to the entire reactor industry.

The fuel-to-moderator ratio of the Maine Yankee reactor (design parameters described in Appendix A) was varied by changing the lattice pitch at fixed fuel pin diameter for a fixed 3.0 w/o initial enrichment. The extrapolated reactivity,

 ρ_0 , as a function of the hydrogen-to-uranium atom ratio (H/U) is plotted in Fig. 7.1. Fig. 7.2 shows a plot of the slope of the reactivity vs. burnup curve as a function of the H/U atom ratio (i.e., A in $\rho = \rho_0 - AB$).

For very wet lattices (H/U > 5.0) the slope is much steeper er than that at the current operating point (H/U = 4.38). The conversion ratio in this range is lower and thus less plutonium is produced and burned. Drying out the lattice raises the conversion ratio and produces more plutonium thereby increasing the net fissile inventory. The presence of the plutonium prevents the reactivity of the lattice from dropping as rapidly as it does in the wetter cases-the slope is less steep.

The conversion ratio continues to increase as the lattice is made drier. In this regime (H/U <2.5), however, the epithermal/thermal flux ratio is quite high and the effective $n(v \sum_{f} / \sum_{a})$ for both U-235 and Pu-239 is reduced. The slope of the reactivity vs. burnup curve $(\frac{d\rho}{dB})$ depends on d ρ , which is a function both of the amount of fissile material produced (hence, conversion ratio) and the reactivity worth of that fissile material. Thus, for these very dry lattices, the decreasing reactivity worth of Pu-239 and U-235 exceeds the benefit of increased Pu-239 production.

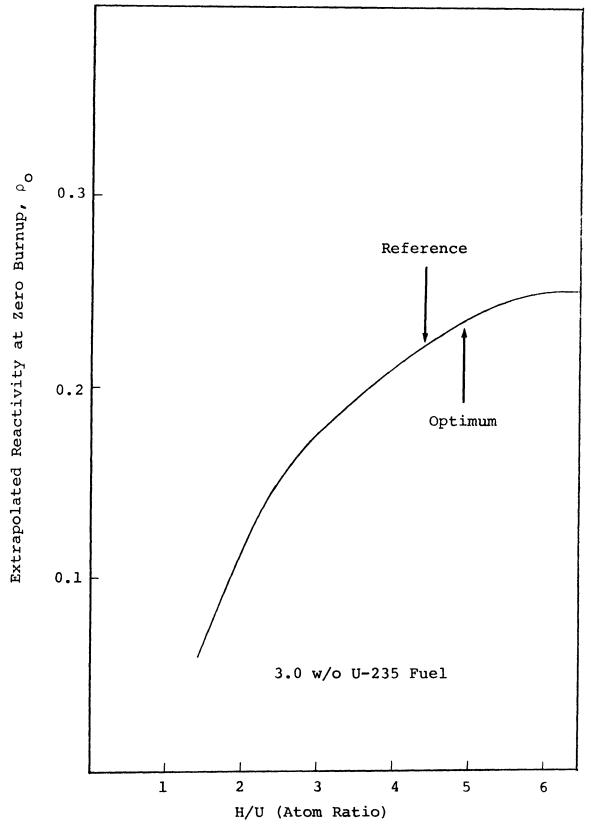


FIG. 7.1 INITIAL EXTRAPOLATED REACTIVITY FOR MAINE YANKEE ASSEMBLIES AS LATTICE PITCH IS VARIED (ε =3.0 w/o)

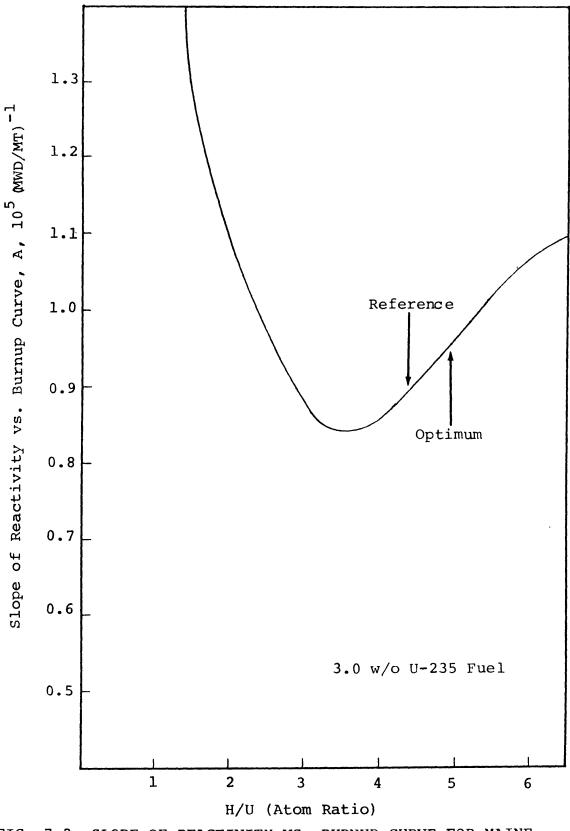


FIG. 7.2 SLOPE OF REACTIVITY VS. BURNUP CURVE FOR MAINE YANKEE ASSEMBLIES AS PITCH IS VARIED (t = 3.0 w/o)

The uranium utilization of the fuel can be written as

$$U_{(\frac{MWD}{MTF})} = \frac{B_{dis}}{(F/P)} = \frac{(\rho_{o} - \rho_{L})}{A_{(\frac{N+1}{2N})} \left(\frac{\varepsilon - 0.2}{0.711 - 0.2}\right)}$$
(7.1)

when equal power sharing among assemblies is assumed. Thus the ratio of ρ_0 to A dominates the behavior of uranium utilization. The uranium utilization of the Maine Yankee reactor as a function of H/U is plotted in Fig. 7.3 for conditions of equal power sharing among assemblies, $\rho_L = 0.042$, and a tails assay of 0.2 w/o U-235. The peak in the uranium utilization curve occurs at 4.85, which is very close to the current assembly operating point (H/U = 4.38) for the Maine Yankee assemblies, and the uranium utilization is not significantly different within this range.

A similar optimization was performed for five batch high discharge burnup fuel (B_{dis} ~50,000 MWD/MT). The curves for extrapolated reactivity (ρ_0), slope (A), and uranium utilization (MWD/MTF) for the 4.34 w/o U-235 lattices as a function of H/U are shown in Figs. 7.4-7.6. The uranium utilization at the current operating point (H/U = 4.38) is 6445 MWD/MTF, whereas at the optimum point on Fig. 7.6 (H/U = ~5.0) the uranium utilization is ~6610 MWD/MTF. This represents a reduction in uranium requirements of about 2.5%, relative to a high burnup lattice operating at the current H/U atom ratio.

It should be noted that for five batch, high burnup cores it may be easier to develop acceptable moderator temperature

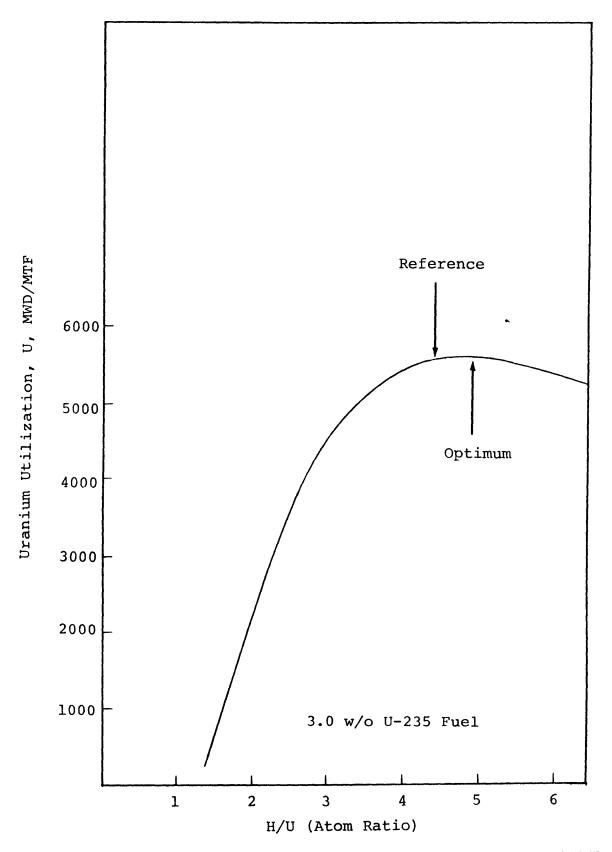


FIG. 7.3 URANIUM UTILIZATION VS. H/U ATOM RATIO FOR MAINE YANKEE ASSEMBLIES AS PITCH IS VARIED (ϵ =3.0 w/o)

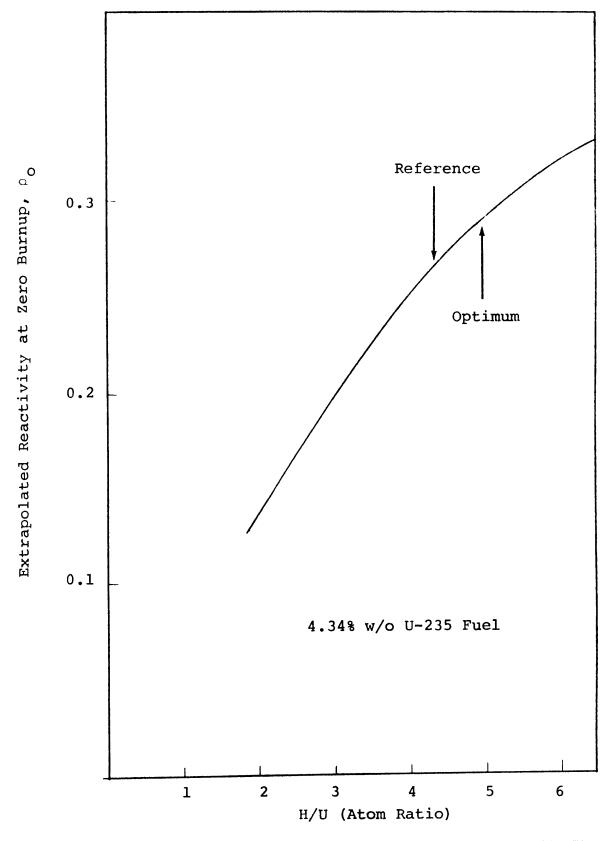


FIG. 7.4 INITIAL EXTRAPOLATED REACTIVITY FOR MAINE YANKEE ASSEMBLIES AS LATTICE PITCH IS VARIED (ε =4.34 w/o)

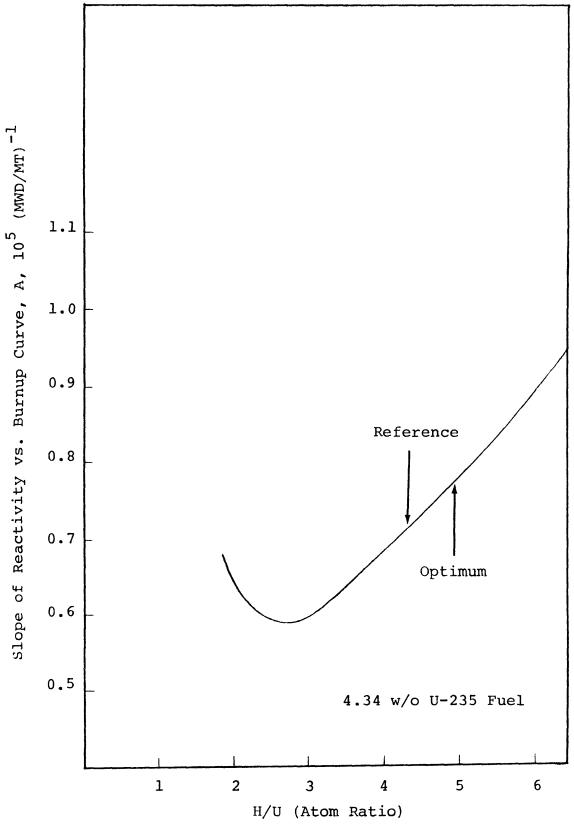


FIG. 7.5 SLOPE OF REACTIVITY VS. BURNUP CURVE FOR MAINE YANKEE ASSEMBLIES AS PITCH IS VARIED (c=4.34 w/o)

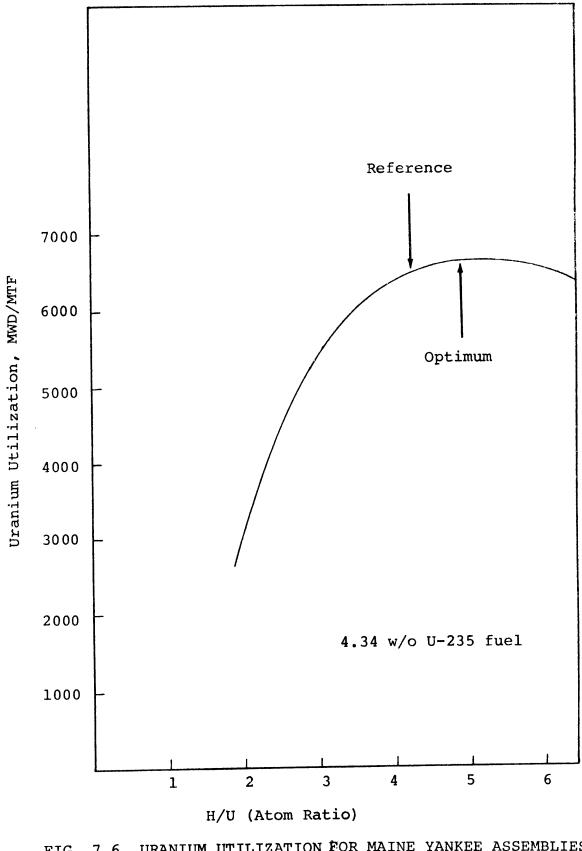


FIG. 7.6 URANIUM UTILIZATION FOR MAINE YANKEE ASSEMBLIES AS PITCH IS VARIED (ϵ =4.34 w/o)

coefficients (M-1) than for three-batch low burnup cases. Thus, the 2.5% reduction in uranium requirements may be achievable without compromising safety margins associated with the moderator temperature coefficient. These benefits can be achieved provided the H/U atom ratio change can be reached without changing lattice pitch or overall reactor size. It must be recognized that this optimization was performed at a fixed density ($\rho = 10.04$ g/cc) and fixed pellet radius (R=0.18225").

The resonance integral of the dominant fertile species (e.g., U-238) plays a key role in the optimization of a lattice with respect to uranium utilization. Isolated-pin effective resonance integrals can be correlated in the form (H-4):

 ${\rm RI}^{28} = {\rm C}_1 + {\rm C}_2 \sqrt{{\rm S/M}}$ (7.2) where ${\rm C}_1$ and ${\rm C}_2$ are constants (which vary slightly depending on the nature of the fuel: oxide, metal or carbide), S is the effective surface area of the fuel pellet and M is the mass of U-238 in the pellet. Since the surface area is proportional to 2IIR_p and the mass of U-238 is proportional to ${\rm IR_p}^2 \rho$, the condition for resonance integral equivalence becomes

$$R_{p} \rho = \text{constant}$$
(7.3)

In practice the situation is more complicated: for example, Dancoff shadowing reduces the effective pin diameter in a tightly-packed lattice. However <u>changes</u> from case to case in the range of practical interest are small. In other words, Eqs. 7.2 and 7.3 can only be used to find the approximate neighborhood of an equivalent set of lattices; the free parameters can then be perturbed until a better match is evidenced

in state-of-the-art computations. Fig. 7.7 shows the relative uranium utilization vs. H/U atom ratio for three cases:

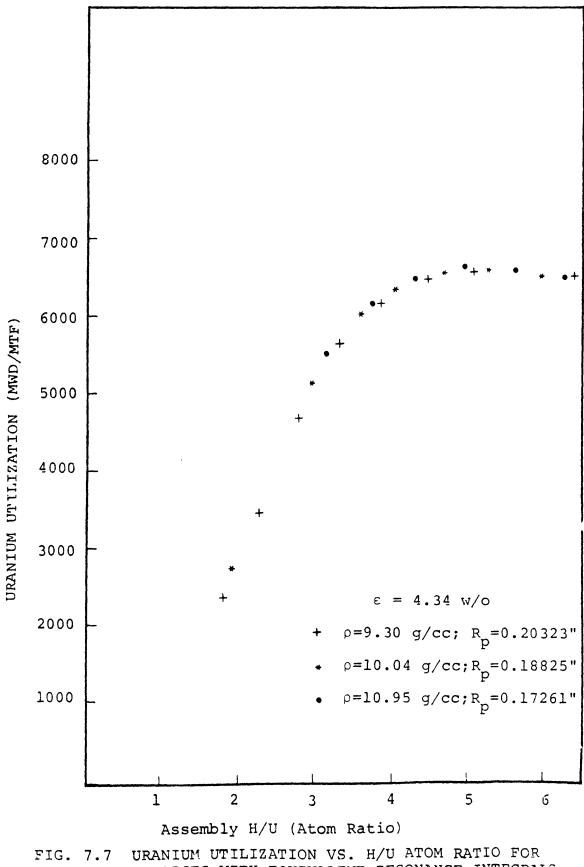
 $\rho = 9.30 \text{ g/cc}; R_p = 0.20323", \rho = 10.04 \text{ g/cc}; R_p=0.18823", \rho = 10.95 \text{ g/cc}; R_p = 0.17261"$

For these cases, $\sqrt{S/M}$ is a constant and, as can be seen from the figure the uranium utilization is, to first order, nearly identical. Differences are less than one percent in the range of interest before any fine tuning. Thus, it appears that a specification of H/U atom ratio and resonance integral is all that is required to determine uranium utilization, inasmuch as their specification determines the value of ρ_0 and A.

It should be recognized, however, that the optimizations shown in Figs. 7.6 and 7.7 were H/U atom ratio optimizations at a fixed value of $\sqrt{S/M}$. A full scope optimization would require a determination of the uranium utilization as a function of both H/U atom ratio and $\sqrt{S/M}$ (for both low burnup and high burnup cases). It is possible that an overall maximum on this three-dimensional surface may occur at an H/U atom ratio and $\sqrt{S/M}$ value different from that shown in Fig. 7.17. If a maximum does exist (and can satisfy the conditions necessary for negative moderator coefficient operation at that point) it may provide additional improvements in uranium utilization.

7.3 Annular Fuel

The use of annular fuel has been proposed as a means to achieve improved uranium utilization, in that it would facilitate higher burnup by allowing more space for fission product gases, and by permitting the fuel to run cooler--thereby



LATTICES WITH EQUIVALENT RESONANCE INTEGRALS

reducing fission product gas release from the UO_2 matrix. Whether or not this is accompanied by inherent physics advantages or disadvantages is still a matter of debate (B-1, M-4, M-5).

Monte Carlo calculations carried out at B&W (B-1) have indicated that at 12% annular voids the use of reduced, uniform pellet smear density to simulate the annular fuel is satisfactory (see Fig. 7.8). Fortunately this is close to the optimum performance point identified by B&W from both a thermal-mechanical and neutronic/economic standpoint. We have therefore used reduced homogeneous pellet smear density to evaluate the use of fuel having a 10% annular void. The most important finding is that the results are sensitive to where one starts and ends the optimization as regards the lattice Table 7.1 summarizes the results for low burnup. H/U ratio. As can be seen LEOPARD pin cell calculations show a slight advantage, while the wetter supercell calculations do not. Wetter pin cells than shown here also exhibited no advantage.

Table 7.2 shows the effects of density changes in uranium utilization for high burnup (ε =4.34 w/o) fuel. It can be seen from the table that a density change at a fixed pellet radius not only changes $\sqrt{\frac{S}{M}}$ but also changes the H/U atom ratio, moving it away from the local optimum defined in Section 7.2.

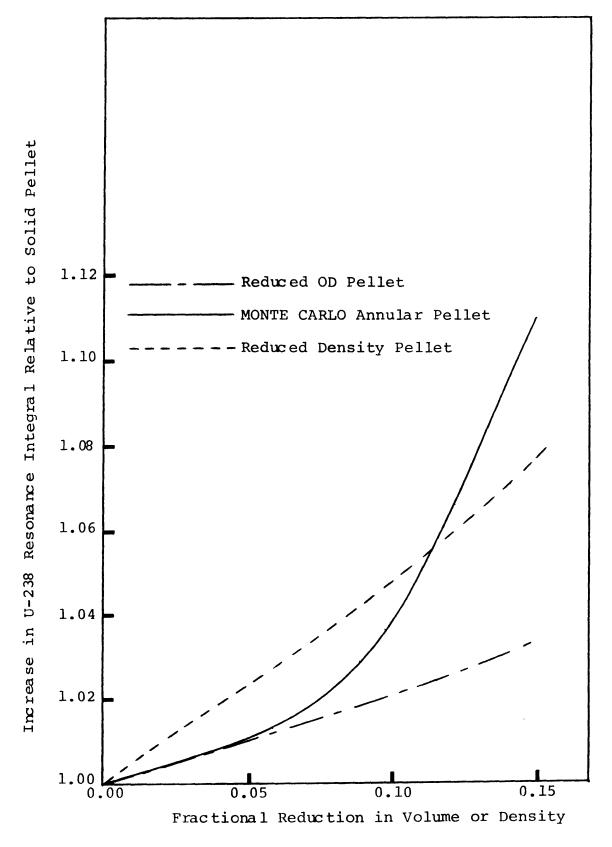


FIG. 7.8 CHANGE IN RESONANCE INTEGRAL AS A FUNCTION OF REDUCTION IN HEAVY METAL CONTENT (B-1)

TABLE 7.1 EVALUATION OF ANNULAR FUEL

Density (g/cc)	Enrichment (w/o U-235)	Core H/U Atom Ratio (HOT,BOL)	MWD MT NAT U FEED		
Leopard Pin Cells					
10.43	2.89	3.448	5129		
10.04	3.00	3.582	5217		
9.40**	3.19	3.826	5347		
9.18	3.30	3.918	5398		
Leopard Super Cells					
10.43	2.89	4.166	5542		
10.04*	3.00	4.381	5540		
9.40**	3.19	4.622	5539		
9.18	3.30	4.733	5549		
Adjusted Super Cells					
10.43	2.89	4.310	5539		
10.04	3.00	4.381	5540		
9.40**	3.19	4.358	5539		
9.18	3.30	4.356	5545		

* Maine Yankee Reference Case

** -Simulates fuel having a 10% annular void.

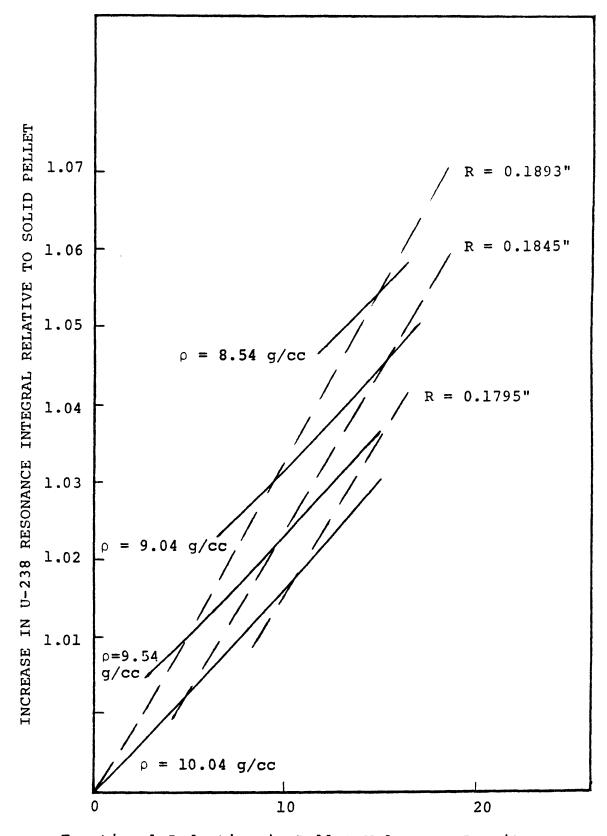
TABLE 7.2	EFFECTS OF FUEL	DENSITY CHANGES	ON URANIUM
	UTILIZATION FOR	HIGH BURNUP FUE	$L (\varepsilon = 4.34)$
	w/o U-235)		

Density (g/cc)	$\sqrt{S/M}$ $(cm^2/g)^{1/2}$	H/U (atom ratio)	Uranium Utilization (MWD/MTF)
10.95	0.9850	4.49	6540
10.40	1.0107	4.72	6557
10.04	1.0287	4.89	6590
9.80	1.0412	5.01	6557
9.30	1.0688	5.28	6540

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The effect of the decrease in the average pellet temperature associated with the use of annular fuel was estimated. To accentuate the effect, the effective resonance temperature was reduced from 1232°F to 700°F in a LEOPARD supercell case. The burnup results indicated a potential reduction in uranium requirements of 0.86% for this extreme temperature perturba-A realistic reduction would be on the order of $20-50^{\circ}$ F tion. depending on design, and thus, the uranium savings would be about 0.03%-0.08% or about 0.00162 %/^OF. To first order, this effect is clearly negligible, particularly since a small change in fuel density or radius can produce comparable effects in non-annular fuel. Thus, we conclude that annular fuel offers no inherent neutronic advantage per se: improvements should be attributed to the concurrent change in S/M and fuel-to-moderator ratio: changes which can also be implemented by other means.

The B&W results shown in Fig. 7.8 might be interpreted to imply that annular fuel enables one to operate at a given H/U ratio with a U-238 resonance integral not achievable by density reduction or fuel rod diameter changes. Fig. 7.9 shows the change in the relative U-238 resonance integral as a function of the fractional reduction in uranium content produced by changes in rod diameter, fuel density, and combinations of the two. It can be seen from the figure that between the bounding curve of fixed density-variable radius (p=10.04 g/cc) and fixed radius-variable density ($R_p = 0.18824$ ")



Fractional Reduction in Pellet Volume or Density FIG. 7.9 RELATIVE U-238 RESONANCE INTEGRAL FOR VARIOUS CHANGES IN PIN DIAMETER AND FUEL DENSITY

any value of the resonance integral within the area can be produced by some combination of rod radius and fuel density (at a given H/U ratio); or, more simply, for any annular fuel design, at a given H/U atom ratio, it is possible to define a neutronically equivalent solid pellet, at the same H/U atom ratio, by manipulating the rod diameter and fuel density.

Despite this fact, there are several possible advantages to annular fuel:

1) The presence of the void offers a compliance volume for increased fission gas production at high burnups, and the cooler fuel releases less gas in the first place.

2) The use of annular fuel may offer a convenient method for reaching a specific H/U ratio without changing rod diameter or lattice pitch, an option which preserves the basic assembly thermal/hydraulic design, which can represent a substantial savings in redesign costs.

3) Annular fuel may be mechanically superior to solid fuel in terms of pellet-clad interactions.

4) There is less stored energy in the fuel and hence it becomes easier to satisfy the ECCS/LOCA (Appendix K) limits.

7.4 Variable Hydrogen/Uranium Atom Ratio

Current pressurized water reactors are designed and the fuel is managed to provide a certain amount of excess reactivity in the core at the beginning of a burnup cycle. This excess reactivity is compensated by soluble boron poison

dissolved in the H_2O moderator. All variable fuel-to-moderator ratio schemes, whether they involve using D_2O in variable amounts to change the H/U atom ratio, or a mechanical movement of the fuel to reduce the effective fuel-tomoderator (and thereby H/U) ratio, are designed to minimize the use of control poison.

At low H/U atom ratios, the fast-to-thermal flux ratio (ϕ_1/ϕ_2) is much higher than that encountered under current PWR operating conditions. As a result, the relative absorption in U-238 is much higher, and the fuel reactivity is decreased. In a sense, these "spectral shift" concepts really trade boron poison for U-238 "poison". The benefit in the spectral shift concept, however, is that the neutron is not lost--the neutron captured in U-238 forms Pu-239, which is fissile. The optimum H/U ratio for a plutonium lattice is even higher than that for a U-235 lattice so that a subsequent "softening" of the spectrum (by increasing the H/U ratio) as the fuel burnup increases, can improve the net uranium utilization of the fuel. The spectral shift concepts have higher conversion ratios (more Pu-239 produced) and potentially better utilization of that plutonium to sustain burnup reactivity lifetime.

Since spectral shift concepts involve eliminating boron control poison, it is useful to examine the maximum theoretical improvement possible if this were accomplished. Recall that the reactivity of a fuel assembly can be accurately

represented by the linear relation

$$\rho = \rho_0 - AB \tag{7.4}$$

where ρ_0 is the extrapolated initial reactivity, A is the slope of the reactivity vs. burnup curve $(\frac{MWD}{MT})^{-1}$, and B is the burnup of the assembly $(\frac{MWD}{MT})$. The slope of the reactivity vs. burnup curve is really the sum of two components:

1) a fission product component which accounts for the essentially linear increase in fission product content with burnup, and

2) a fissile burnup component which accounts for the net decrease in the fissile material inventory with burnup.

Thus, the reactivity of a fuel assembly can be written as:

$$\rho = \rho_{o} - A_{FP} B - A_{burn} B$$
(7.5)

where A_{FP} is the component of the slope caused by fission product buildup, A_{burn} is the component of the slope caused by fissile inventory decrease, and A_{burn} plus A_{FP} equals A. Results from LEOPARD supercell calculations run with full fission product inventories and with zero fission product inventories give:

$$\rho_{0} = 0.220804$$

$$A_{total} = 0.89242 \times 10^{-5} \left(\frac{MWD}{MT}\right)^{-1}$$

$$A_{FP} = 0.55385 \times 10^{-5} \left(\frac{MWD}{MT}\right)^{-1}$$

$$A_{burn} = 0.33857 \times 10^{-5} \left(\frac{MWD}{MT}\right)^{-1}$$

for a 3.0 w/o U-235 initial enrichment. The discharge burnup of the fuel can now be written as:

$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A_{FP} (\frac{N+1}{2N}) + A_{burn} (\frac{N+1}{2N})}$$
(7.6)

where N is the number of in-core batches, $\rho_{\rm L}$ is the leakage reactivity, and equal power sharing among assemblies has been assumed. For the case of on-line refueling (N = ∞) boron is not necessary because the excess reactivity of the freshest fuel is balanced by the reactivity deficit of the oldest fuel. Neutrons are shared internally among fuel assemblies and none are lost to boron. Such neutron sharing can be imitated by absorbing neutrons in U-238 to form Pu-239 and subsequently burning the Pu-239 later in life. The neutron is, in a sense, "stored" until it is needed. For N-batch cores (N < ∞) with optimal spectral shift (no boron absorptions) the discharge burnup can be estimated by taking the continuously renewed fuel burnup limit of Eq. 7.6:

$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A_{FP} (\frac{N+1}{2N}) + A_{burn} (\frac{\omega+1}{2\omega})}$$
$$= \frac{\rho_{o} - \rho_{L}}{A_{FP} (\frac{N+1}{2N}) + \frac{1}{2} A_{burn}}$$
(7.7)

Table 7.3 lists the estimated discharge burnups for normal boron poison control and spectral shift control, and the uranium requirement benefit as a function of the number of in-core batches for a fixed radial and axial leakage of 0.042.

3 ESTIMATED THEORETICAL BENEFITS OF OPTIMIZED SPECTRAL SHIFT CONTROL AS A FUNCTION OF THE NUMBER OF BATCHES

itches	Normal Burnup (MWD/MT)	Spectral Shift Burnup Fission Product N = N Fuel Burnup N = ∞ (MWD/MT)	Percent Decrease in Uranium Requirement
	20044	29050	31.1
	26726	33682	21.7
	30067	35573	15.5
	32071	36600	12.4
	33407	37246	10.3
	36444	38607	5.6
	40088	40088	0.0

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The relative benefits of spectral shift control decrease as the number of staggered fuel reload batches increases. Boron control requirements are lower for an increased number of batches, thus, elimination of boron has a correspondingly reduced advantage.

The LEOPARD code contains an option which permits one to progressively vary the D₂O to H₂O ratio in a unit cell burnup calculation. A slight modification to the code was made to permit the inclusion of voids (instead of D_2O) in the H_2O moderator. Thus, mechanical spectral shift concepts could be evaluated without the complication of D₂O scattering and moderation. The case of single-batch fuel management, no leakage, and continuous mechanical spectral shift was consid-The void content input into LEOPARD was adjusted such ered. that the unit cell was held just critical at all points. Eq. 7.5 (applied to the reference supercell) predicts a discharge burnup of 34473 MWD/MT (for zero leakage), whereas the LEOPARD results using the continuous spectral shift technique, show a discharge burnup (for zero leakage) of about 34270 MWD/MT for a return to the same (BOL) H/U ratio (4.38) as the reference supercell. Further increases in H/U atom ratio (H/U >4.38) will increase the reactivity of the lattice and extend the burnup of the single batch case by

(approximately) an additional 2500 MWD/MT. There are several points which must be noted regarding these results:

1) The optimum H/U atom ratio of the fuel is a burnupdependent function which also depends on the time-dependent path that the lattice takes in H/U "phase space".

2) The fission products will, in general, have a different reactivity worth in the spectral shift cases relative to the fixed H/U designs. This raises some question as to the applicable value of $A_{\rm FP}$ in Eq. 7.5. Perhaps determination of $A_{\rm FP}$ at some cycle-average, or endpoint H/U atom ratio would be more appropriate.

3) A continuous spectral shift applied to high discharge burnup fuels (high initial enrichment) would produce more plutonium and may yield better uranium savings.

4) The accuracy of the LEOPARD results are open to question for very dry lattices (C-1).

Based on these considerations, the results of Table 7.3 should be considered to be at least slightly optimistic with respect to the predicted physics benefits. Engineering constraints present additional problems:

1) Accomplishing the single-batch spectral shift described via mechanical means would probably be impractical from a thermal/hydraulic standpoint in current reactor designs because of the attendant degradation in heat transfer conditions for the farthest off-design configurations.

2) Many concepts which might permit a continuously variable fuel-to-moderator ratio would probably introduce safety

problems because of the possibility of accidental reactivity insertions.

3) The more practicable mechanical spectral shift techniques would require extensive manual fuel assembly adjustments during refueling outages and would therefore be noncontinuous in nature. The introduction of such "discretization" would certainly reduce the maximum achievable benefits of spectral shift. Thus, Table 7.3 should be considered extremely optimistic with respect to thermal/hydraulic and mechanical considerations.

7.5 H/U Atom Ratio Changes During Refueling

Various schemes have been proposed for accomplishing mechanical spectral shift in PWRs; most involve pulling pins from assemblies to produce "wetter" lattices and the formation of new assemblies from the pins that were removed (R-1). Figure 7.10 shows one such scheme. Fresh fuel is loaded (53 assemblies with 264 pins/assembly) in a reactor with 193 assemblies. After cycle one, during the refueling interval, 64 pins are pulled from each fresh assembly and 17 additional fuel assemblies are manufactured. Thus, the cycle two fuel is "wetter" than the cycle one fuel, and there are more assemblies of type two fuel than there are of type one fuel. From a neutronic standpoint, the same type of "dry-wet" spectral shift can be accomplished by the scheme shown in Fig. 7.11. In this scheme, the cycle one fuel operates at the same H/U atom ratio as the cycle one fuel of the previous

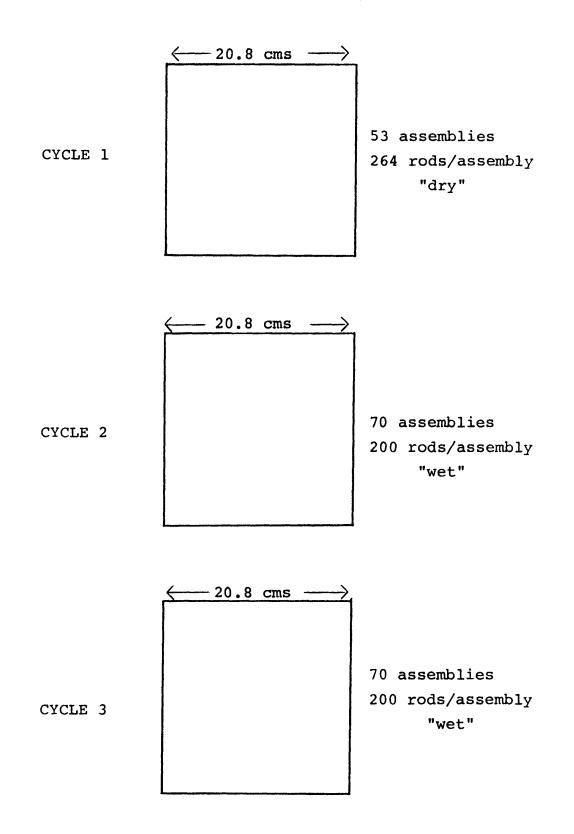


FIG. 7.10 LATTICE SPECTRAL SHIFT SCHEME USING PIN-PULLING AND BUNDLE RECONSTITUTION

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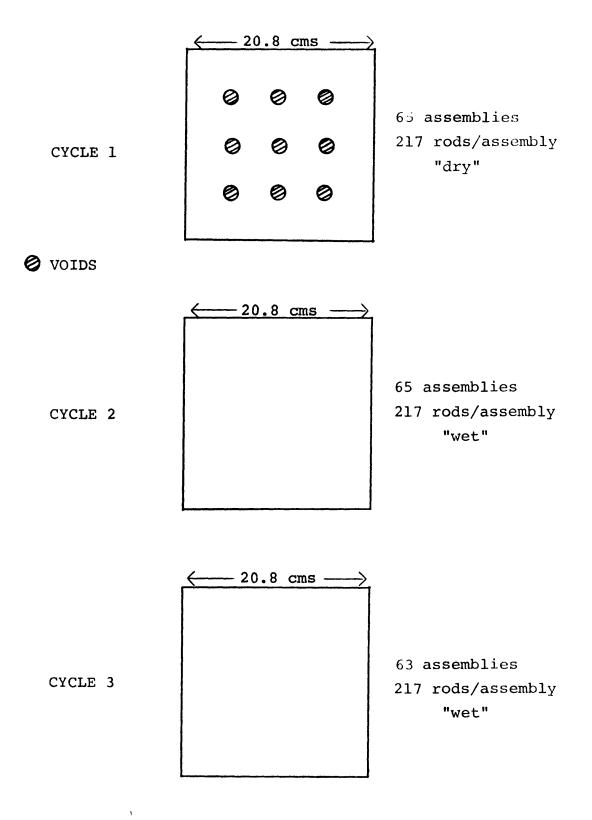


FIG. 7.11 LATTICE SPECTRAL SHIFT SCHEME USING "VOID" INCLUSION IN CYCLE ONE ASSEMBLIES scheme, but the same H/U atom ratio is produced by introducing voids into the coolant (in some unspecified fashion). At the refueling interval, the voids are removed and the lattice is as wet as the type two fuel from the previous scheme. In this scheme, the number of pins per assembly and the number of assemblies per batch remains constant.

It is shown in Appendix G that in terms of the uranium utilization, these two schemes are absolutely equivalent. However, it is easier to calculate the optimum uranium savings for the scheme shown in Fig. 7.11 since the complications associated with feed adjustments and power sharing become more transparent for this evaluation. The unit cell calculations were made with LEOPARD, using the variable void option programmed into the code (as discussed in Appendix B). (It is not clear that such a variable void scheme could ever be practical). Lattice modifications must be made during refuelings and the time between refuelings must remain sufficiently long that the overall capacity factor is not adversely affected. The fact that the lattice modifications remain fixed during the cycle reduces the flexibility (and potential benefits) of this scheme. Two types of fuel management scenarios have been developed for mechanical spectral shift concepts (R-1). In Concept 1, the fresh fuel lattice is drier than the reference case at the beginning of a cycle and the reactor is run until it reaches criticality with no boron present. The oldest fuel batch is discharged, fresh fuel is loaded, and the lattice modifications are made to the once burned fuel.

In Concept 2, the fresh fuel is drier than the reference case at beginning-of-cycle and the reactor is run until it reaches criticality with no boron present. The reactor head is removed and lattice modifications are made to the freshest fuel (which inserts reactivity) without discharging the oldest fuel batch. The vessel is then closed and the reactor is restarted and run until the extra reactivity insertion is burned out.

The reactivity vs. burnup curves for the procedures of interest are diagrammed in Figs. 7.12-7.14. The first figure represents normal three batch fuel management. Batches 1-3 start irradiation at points A_1 , A_2 and A_3 , and complete irradiation at EOC at points B_1 , B_2 and B_3 . Concept 1 is illustrated in Fig. 7.13. The batches begin irradiation at points C_1 , C_2 , and C_3 and end irradiation at D_1 , D_2 and D_3 . During the refueling outage batch three is discharged and the lattice of batch one (which becomes batch two in the new cycle) is adjusted such that reactivity is inserted. The reactivity change ($\Delta \rho$) is given by

$$\Delta \rho = \rho(C_2) - \rho(D_1) \tag{7.8}$$

It is important to recognize that in normal (non-spectral shift, fixed lattice) fuel management the reactivity was considered a "state function" or "state vector function" of the burnup. However, for spectral shift designs the reactivity is now a path-dependent function, i.e. the reactivity at any point depends not only on the burnup but also on the path

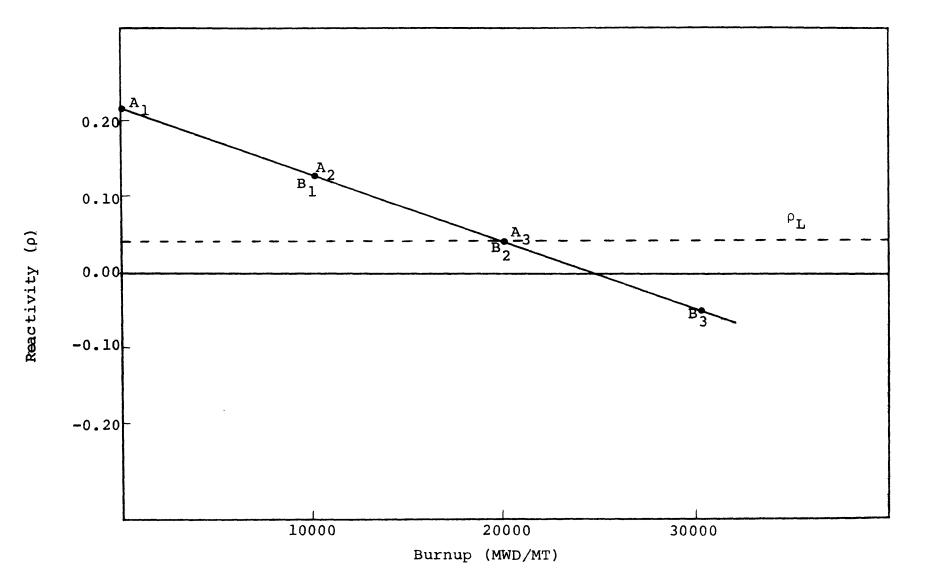


FIG. 7.12 REACTIVITY VS. BURNUP FOR CONVENTIONAL THREE BATCH FUEL MANAGEMENT

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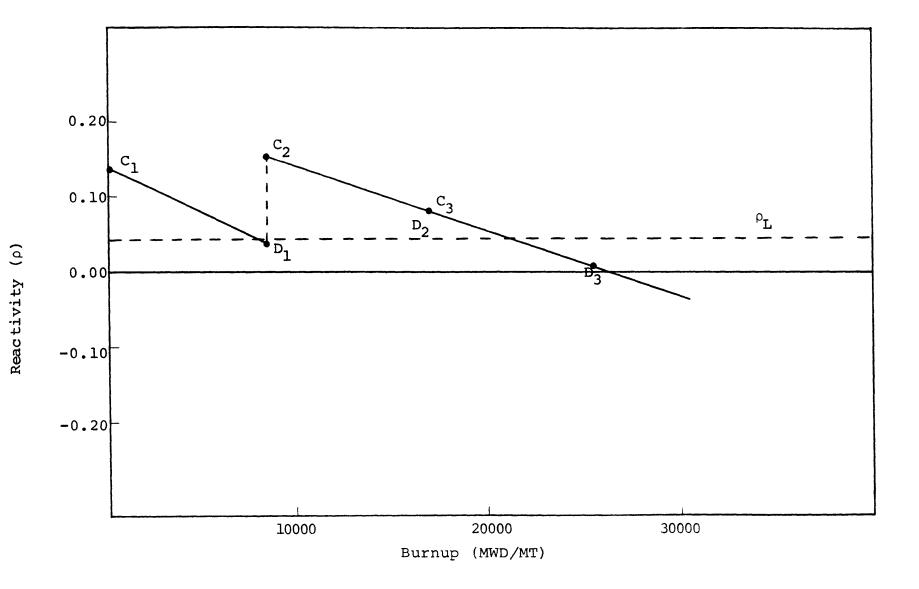


FIG. 7.13 REACTIVITY VS. BURNUP FOR CONCEPT 1 MECHANICAL SPECTRAL SHIFT

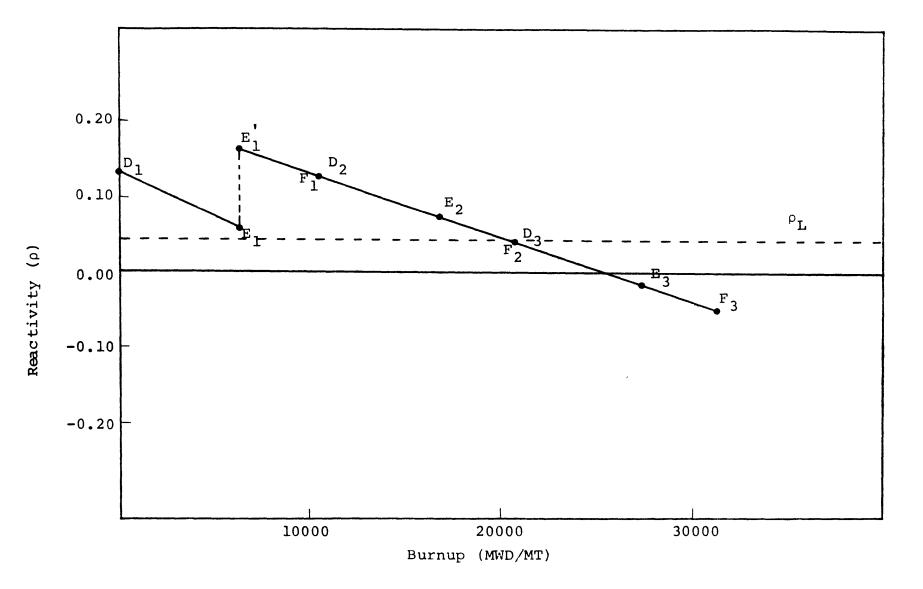


FIG. 7.14 REACTIVITY VS. BURNUP FOR CONCEPT 2 MECHANICAL SPECTRAL SHIFT

taken (in H/U phase space) to achieve that burnup. Thus, the points C_1 , C_2 , C_3 , D_1 , D_2 and D_3 , and $\Delta\rho$ may vary significantly depending on the H/U ratios the assembly has encountered during burnup. It is <u>critical</u> to recognize that these points cannot be arbitrarily determined once the H/U atom ratio changes have been prescribed.

A possible reactivity vs. burnup curve for a Concept 2 design is shown in Fig. 7.14. The fuel batches begin the cycle at the points D_1 , D_2 and D_3 . Burnup continues until criticality is reached at points E_1 , E_2 and E_3 . The reactor head is removed and the lattice of batch 1 is adjusted to insert reactivity (point E'_1). The vessel is closed (without removing fuel assemblies) and the reactor is restarted and operated until the excess reactivity, $\Delta \rho (E'_1 - E_1)$, is consumed. The oldest fuel is discharged at point F_3 .

Variable lattice schemes can be generalized to any number of batches and/or reactivity insertions (lattice changes) at any of several points in the life of the fuel assembly. Although no attempt was made to find the optimal H/U (atom ratio) trajectory, a methodology based on the Advanced Linear Reactivity Model was developed, which is general enough to include any realistic combination of lattice changes and batch fractions, for the calculation of uranium utilization.

Recall that for ordinary three batch fuel management the reactivity equations of the batches can be written as:

$$\rho_{1} = \rho_{0} - AB_{dis} f_{1}$$

$$\rho_{2} = \rho_{0} - AB_{dis} (f_{1} + f_{2})$$

$$\rho_{3} = \rho_{0} - AB_{dis} (f_{1} + f_{2} + f_{3})$$
(7.9)

and the condition for criticality is:

$$\rho_{\rm L} = \sum_{i=1}^{3} f_i \rho_i \qquad (7.10)$$

which gives:

$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A\left\{\begin{array}{c}3 & 3\\ \sum & \sum\\i=1 & j=i\end{array}\right\}}$$
(7.11)

Consider the Concept 1 design shown in Fig. 7.13. The reactivity equations can be written as:

$$\rho'_{1} = \rho'_{0} - A'B_{dis} f_{1}$$

$$\rho'_{2} = \rho'_{0} - A'B_{dis} f_{1} + \Delta \rho - A''B_{dis} f_{2}$$

$$\rho'_{3} = \rho'_{0} - A'B_{dis} f_{1} + \Delta \rho - A''B_{dis} (f_{2} + f_{2})$$
(7.12)

In this set of equations ρ'_{O} is the initial extrapolated reactivity of the lattice corresponding to point C_{1} , A' is the slope of the reactivity vs. burnup curve during the first cycle, $\Delta \rho$ is the reactivity insertion from lattice changes $(\rho(C_{2}) - \rho(D_{1}))$, and A" is the slope of the reactivity vs. burnup curve after the lattice change. Applying the criticality condition of Eq. 7.10 gives:

$$B_{dis} = \frac{\rho_{o}' - \rho_{L} + (f_{2} + f_{3}) \Delta \rho}{A' \left[f_{1}^{2} + f_{1} f_{2} + f_{1} f_{3} \right] + A'' \left[f_{2}^{2} + f_{2} f_{3} + f_{3}^{2} \right]}$$
(7.13)

For a fuel rod with a specific value of $\sqrt{S/M}$, ρ'_{O} and A' are functions of the initial H/U atom ratio, A" depends on the final H/U atom ratio, and $\Delta\rho$ depends on the transition ("drywet") burnup B_{tran}. Eq. 7.13 must be optimized with respect to power split (f₁, f₂, f₃). Eq. 7.13 can be rewritten as:

$$B_{c} = \frac{\rho_{o}' - \rho_{L} + (1 - f_{1}) \Delta \rho}{3A'f_{1} + 3A'' \left[f_{2}^{2} + f_{2}f_{3} + f_{3}^{2}\right]}$$
(7.14)

The denominator is optimized (minimized) when $f_2 = f_3$ hence: $B_c = \frac{\rho_0' - \rho_L + (1-f_1) \Delta \rho}{3A'f_1 + 3A'' 3(\frac{1-f_1}{2})}$ (7.15)

Since the lattice modification is made at the refueling, ${}^{3B}_{c} f_{1} = B_{tran}$ and hence:

$$B_{tran} = \frac{f_1 \rho_0' - \rho_L + (1 - f_1) \Delta \rho}{A' f_1 + A'' 3(\frac{1 - f_1}{2})}$$
(7.16)

Thus, for Concept 1 fuel schemes, the choice of transition burnup, initial H/U atom ratio and final H/U atom ratio specifies the discharge burnup, i.e., Eq. 7.16 specifies f_1 absolutely. Concept 2 schemes are somewhat more difficult to analyze because they involve an extension of cycle using the same fuel. If the cycle length between points D_i and E_i in Fig. 7.14 (i = 1,2,3) is designated as B_c and the span from points E_i to F_i (i = 1,2,3) designated as AB_c , then the reactivity equations can be written for B_c as:

$$\rho_{1}^{\prime} = \rho_{0}^{\prime} - A^{\prime}B_{c} N f_{1}$$

$$\rho_{2}^{\prime} = \rho_{0}^{\prime} - A^{\prime}B_{c} N f_{1} + \Delta \rho - A^{"}\Delta B_{c} N f_{1}^{*} - A^{"}B_{c} N f_{2}$$

$$\rho_{3}^{\prime} = \rho_{0}^{\prime} - A^{'}B_{c} N f_{1} + \Delta \rho - A^{"}\Delta B_{c} N f_{1}^{*} - A^{"}B_{c} N f_{2}$$

$$- A^{"}\Delta B_{c} N f_{2}^{*} - A^{"}B_{c} N f_{3} \qquad (7.17)$$

and the equations for ΔB_{c} can be written as:

$$\rho_{1}^{\prime} = \rho_{0}^{\prime} - A^{\prime}B_{C} Nf_{1}^{\prime} + \Delta \rho - A^{\prime\prime}\Delta B_{C} N f_{1}^{\star}$$

$$\rho_{2}^{\prime} = \rho_{0}^{\prime} - A^{\prime}B_{C} N f_{1}^{\prime} + \Delta \rho - A^{\prime\prime}\Delta B_{C} N f_{1}^{\star} - A^{\prime\prime}B_{C} N f_{2}^{\prime}$$

$$- A^{\prime\prime}\Delta B_{C} N f_{3}^{\star}$$

$$\rho_{3}^{\prime} = \rho_{0}^{\prime} - A^{\prime} B_{C} N f_{1}^{\prime} + \Delta \rho - A^{\prime\prime}\Delta B_{C} N f_{1}^{\star} - A^{\prime\prime} B_{C} N f_{2}^{\prime}$$

$$- A^{\prime\prime}\Delta B_{C} N f_{2}^{\star} - A^{\prime\prime} \Delta B_{C} N f_{3}^{\star} - A^{\prime\prime} B_{C} N f_{3}^{\star} \qquad (7.18)$$

where ρ_{O}^{\prime} is the reactivity at point D_{1} , $\Delta \rho$ is the reactivity change $(\rho(E_{1}^{\prime}) - \rho(E_{1}))$, A' is the slope of the reactivity vs. burnup curve before lattice modification, A" is the slope of the reactivity vs. burnup curve after lattice modification, N is the number of batches, B_{C}^{\prime} is the cycle length during irradiation between points D_{i} and E_{i} , ΔB_{C}^{\prime} is the cycle length for irradiation between points E_{i}^{\prime} and F_{i}^{\prime} , f_{i}^{\prime} is the power fraction in batch i at the end of B_{C}^{\prime} and $f_{i}^{\star}^{\star}$ is the condition for Eq. 7.17 is:

$$\rho_{\rm L} = \sum_{i=1}^{3} f_i \rho_i \qquad (7.19)$$

whereas the criticality condition for Eq. 7.18 is:

$$\rho_{\mathbf{L}} = \sum_{\mathbf{i}=1}^{3} \mathbf{f}_{\mathbf{i}}^{*} \quad \rho_{\mathbf{i}}$$
(7.20)

Equations 7.17 to 7.20 can be combined to yield

$$B_{c} = \frac{\int_{0}^{t} -\rho_{L} + \Delta\rho(f_{2} + f_{3}) - \left\{\frac{\int_{1}^{2} \Delta\rho\left\{f_{1}^{*}f_{2} + f_{1}^{*}f_{3} + f_{2}^{*}f_{3}\right\}}{\left\{\sum_{i=1}^{3} \int_{j=i}^{3} f_{i}^{*}f_{j}^{*}\right\} - (f_{1}^{*}f_{2} + f_{1}^{*}f_{3} + f_{2}^{*}f_{3})} - \left(f_{1}^{*}f_{2} + f_{1}^{*}f_{3} + f_{2}^{*}f_{3}\right)}{\left\{\sum_{i=1}^{3} \int_{j=i}^{3} f_{i}^{*}f_{2}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}f_{2}^{*}-f_{2}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{2}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{2}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{2}^{*}f_{3}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{2}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{3}^{*}-f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{3}^{*}-f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{3}^{*}-f_{3}^{*}f_{3}^{*}-f_{2}^{*}f_{3}^{*}-f_{3}$$

and:

$$\Delta B_{c} = \frac{f_{1}\Delta \rho - A'' B_{c} N \left[f_{2}f_{2}^{*} + f_{2}f_{3}^{*} + f_{2}f_{3}^{*} - f_{2}^{2} - f_{2}f_{3} - f_{3}^{2} \right]}{A'' N \left[\left\{ \sum_{i=1}^{3} \sum_{j=i}^{3} f_{i}^{*}f_{j}^{*} \right\} - f_{1}^{*}f_{2} - f_{1}^{*}f_{3} - f_{2}^{*}f_{3} \right]}$$
(7.22)

These two equations must be solved for B_c and ΔB_c . There are, however, six additional unknowns, the f_i and f_i^* . The additional equations required are:

$$f_1 + f_2 + f_3 = 1$$
 (7.23)

$$f_1^* + f_2^* + f_3^* = 1$$
 (7.24)

$$B_{c} N f_{l} = B_{tran}$$
(7.25)

The power fractions can be estimated by

$$f_1 = C_1 f_1^*$$
 (7.26)

$$f_2 = C_2 f_3$$
 (7.27)

$$f_2 = C_3 f_3^*$$
 (7.28)

where C_1 , C_2 and C_3 are constants (~1.0). The solution of Eqs. 7.21 and 7.22 using the conditions of Eqs. 7.26-7.28 gives the discharge burnup for a Concept 2 fuel management scheme which involves a lattice change from a specific initial H/U atom ratio to a final H/U atom ratio at a given transition burnup, B_{tran} . To determine the discharge burnup for more realistic power schedules, the conditions of Eqs. 7.26-7.28 must be adjusted. The optimum burnup schedule can be calculated by setting $C_2 = C_3 = 1.0$ and by iteratively solving for C_1 until $B_c + \Delta B_c$ is a maximum.

In realistic cases, the power split among assemblies should be determined by the k^{Θ} method developed previously and hence, the optimum burnup may not be achievable in a practical case.

Table 7.4 shows the results of a preliminary optimization of the Maine Yankee reactor assembly design for the Concept 1 fuel management scheme for a 3.0 w/o U-235 enr.chment. None of the cases analyzed show any decrease in u anium requirements.

Table 7.5 shows the results for the same LEOPARD cases using the Concept 2 fuel management scheme. The conditions used were $f_2 = f_3 = f_2^* = f_3^*$ and $f_1 = f_1^*$. The values of f_1 in the converged solution are shown for each case. The optimum point occurs at an H/U atom ratio shift from 2.936 to 4.894.

The value of cycle burnup for this "best" lattice change corresponds to a choice of $C_1 = 1.0$ or $f_1 = f_1^*$. The value of C_1 was varied to determine if further benefits could be realized. At a value of $C_1 = 0.8919$ corresponding to $f_1 = 0.4126$, $f_1^* = 0.4626$, $B_c + \Delta B_c$ was 11089 MWD/MTP. This power history/power map is somewhat unrealistic; however, if it could be achieved a 9% reduction in uranium requirements would be realized.

Tables 7.4 and 7.5 do not represent a complete optimization of the Maine Yankee design. It may be possible that at other rod diameters or pellet densities (different $\sqrt{S/M}$) H/U atom ratio shifts may exist which would give improvements for both Concept 1 and Concept 2 schemes. Since the power split among assemblies is governed by the value of k and 0 (as shown in Chapter 4) a thorough optimization would include

Initial H/U (HOT, BOL)	FINAL H/U (HOT, Relative to U at BOL)	Bc	Relative Feed Requirement
4.381*	4.381	10085	1.000
2.551	3.644	8585	1.175
2.190	4.381	8651	1.166
2.629	4.381	9214	1.095
3.067	4.381	9523	1.059
3.505	4.381	9686	1.041
3.943	4.381	9770	1.032
2.447	4.894	9214	1.095
2.936	4.894	9 539	1.057
3.426	4.894	9736	1.036
3.915	4.894	9872	1.022
2.780	5.560	9444	1.078
3.336	5.560	9605	1.050
3.892	5.560	9815	1.028

TABLE 7.4 RELATIVE URANIUM FEED REQUIREMENTS FOR VARIOUS CONCEPT 1 STRATEGIES (B = 8500 MWD/MTP)

INITIAL H/U (HOT, BOL)	FINAL H/U (HOT, Relative to U at BOL)	$\underline{\mathbf{B}_{\mathbf{c}}^{+ \wedge \mathbf{B}}}_{\mathbf{c}}$	<u>f</u> 1	REL. FEED
4.381	4.381	10085	0.33	1.000
2.551	3.844	9917	0.46	0.983
2.629	4.381	10786	0.42	0.935
3.067	4.381	10535	0.34	0.957
3.505	4.381	10350	0.32	0.974
3.943	4.381	10067	0.30	1.002
2.447	4.894	10509	0.59	0.950
2.936	4.894	10979	0.38	0.919
3.426	4.894	10640	0.33	0.948
3.915	4.894	10340	0.30	0.975
2.780	5.560	10728	0.34	0.940
3.336	5.560	10468	0.31	0.963
3.892	5.560	10289	0.30	0.980

TABLE 7.5	RELATIVE URANIUM FEED	REQUIREMENTS	FOR VARIOUS
	CONCEPT 2 STRATEGIES ($B_{tran} = 8500$	MWD/MTP)

these effects in the calculation of initial H/U, final H/U, $\sqrt{S/M}$, and transition burnup B_{tran} . To first order, it appears that Concept 1 offers no uranium savings (and perhaps even some losses) and Concept 2 (when well-optimized) may provide uranium savings on the order of 10%. A complete optimization would be required to maximize Concept 2 savings. However, the results of such an optimization may suggest lattice changes that would violate thermal-hydraulic margins in the reactor and thus, these reductions in uranium requirements may not be achievable in actual practice.

7.6 Chapter Summary

Optimization of the Maine Yankee reactor design for H/U atom ratio suggests that at current discharge burnups (<36000 MWD/MT) there is no benefit to changing the current lattice design. On the other hand, at higher discharge burnups (~50000 MWD/MT) a potential 2.5% reduction in uranium requirements relative to the current H/U atom ratio can be achieved by a slight wettening of the lattice.

The use of annular fuel does not appear to offer any uranium savings per se, but annular fuel may facilitate operation, in a mode (wetter, higher burnup) which does improve uranium utilization. LEOPARD results indicate that for any annular fuel it is possible to find a solid pellet at an appropriate rod diameter and fuel density that is neutronically equivalent.

A methodology using the Advanced Linear Reactivity Model has been developed which can evaluate mechanical spectral shift concepts on a self-consistent basis. Mechanical spectral shift techniquesbased on the Concept 1 scheme (dry assemblies cycle 1, wet assemblies for cycles 2 and 3) do not appear to offer any uranium savings. It must be noted, however, that a thorough optimization was not performed. Nevertheless, all the cases analyzed using this scheme failed to break even with the reference case uranium utilization.

The Concept 2 scheme (dry-wet spectral shift <u>during</u> cycle 1), on the other hand, did produce uranium savings in some cases. The best case identified produced a reduction in uranium requirements of ~9%. Again, it must be noted that a thorough optimization may produce larger uranium savings.

For Concept 2 cases, however, the number of refueling shutdowns is doubled and this will adversely affect overall reactor capacity factors.

It must be conceded, however, that if a viable scheme could be devised for <u>continuously</u> varying lattice water content during normal operation, substantial improvements in uranium utilization could be realized (~18% for threebatch cores; ~10% for five-batch cores). So far, however, no one has come forth with an acceptable (economic, failsafe) means to attain this goal other than the often considered (and rejected) H_2O/D_2O spectral shift concept.

CHAPTER 8

SUMMARY, CONCLUSIONS, AND RECOMMENDATIONS

8.1 INTRODUCTION

Over the past several years both political and economic considerations have precluded the commercial reprocessing of nuclear fuels in the United States. In the absence of reprocessing and breeder reactors it is desirable to extend uranium resources by optimizing current generation LWRs to use uranium on the once-through fuel cycle as efficiently as possible. Since the majority (nearly 2/3) of LWRs in the United States are pressurized water reactors, the research reported here has focused on the improvement of uranium utilization in current generation PWRs operating in the once-through mode. In addition, the emphasis has been on changes which can be readily backfit into existing reactors.

An important part of the present work has been the concern with the formulation of simple, but accurate models which can be used to separate out the concurrent contributions of simultaneous changes in several controlling variables (e.g. power history/power map and some design change of interest). Thus every effort is made to avoid obscuring the effect of concern, and making comparisons on an "all else being equal" basis.

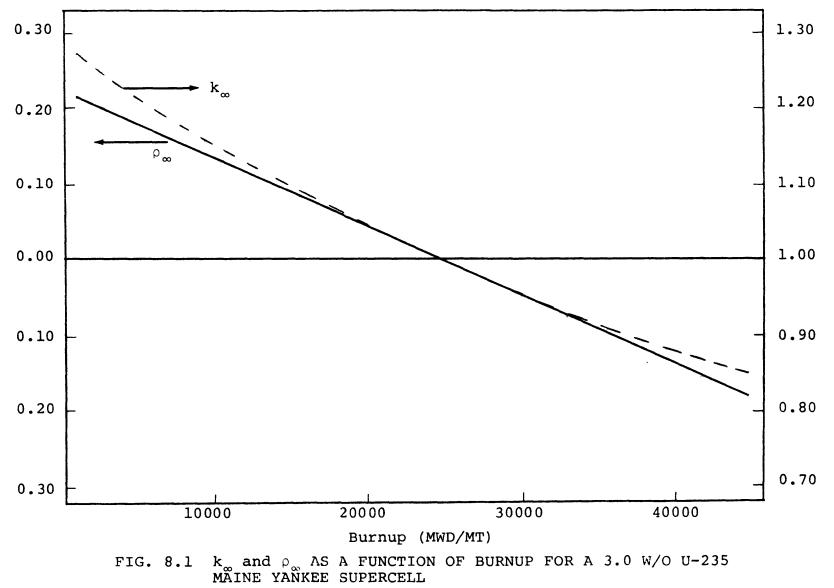
8.2 BACKGROUND AND RESEARCH OBJECTIVES

The work described in this report was undertaken under the LWR Technology Program for Improved Uranium Utilization sponsored by the United States Department of Energy. Many investigators, both at MIT and elsewhere, have provided input along the same general lines to the Nonproliferation Alternative Systems Assessment Program (NASAP) (N-1), which in turn, provided input to the International Nuclear Fuel Cycle Evaluation (INFCE) (I-1). These investigators identified a variety of uranium-saving techniques which deserved further scrutiny

Consequently the general goal of the present work has been to develop, test, and apply self-consistent, methods for evaluating improvements in core design and fuel management strategy which would increase uranium utilization in PWRs.

8.3 Methodology Development

In order to analyze complex fuel management strategies the "Advanced Linear Reactivity Model" was developed. It was recognized that in the current range of interest with respect to fuel lattice parameters, reactivity, ρ , as a function of burnup, B ($\frac{MWD}{MT}$), was extremely linear, in fact, substantially more linear than the multiplication factor, k, as a function of burnup (See Fig. 8.1). Further analysis showed that the preferred



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scheme for the calculation of the reactivity of combinations of assemblies in a PWR was power fraction weighting; namely,

$$\rho_{\text{sys}} = \sum_{i=1}^{N} f_{i} \rho_{i}$$
(8.1)

This formulation was verified (maximum error <0.4%) using PDQ-7. It was next shown that the effects of neutron leakage from the core could be incorporated as a decrement to system reactivity:

$$\rho_{L} = \frac{A_{ex-core}}{A_{total} k_{sys}} = \frac{A_{ex-core}}{F_{total}}$$
(8.2)

where $\rho_{\rm L}$ is the leakage reactivity, $A_{\rm ex-core}$ is the total absorption in non-core material (i.e. shroud, barrel, ex-core H₂O), $A_{\rm total}$ is the total absorption in the reactor (core + non-core material), $k_{\rm sys}$ is the system multiplication factor and $F_{\rm total}$ is the total neutron production in the reactor. Hence, the criticality condition in a reactor becomes:

$$\rho_{\mathbf{L}} = \sum_{i=1}^{N} f_{i} \rho_{i}$$

$$(8.3)$$

For an N-batch core, the reactivity vs. burnup equations can be written as:

$$\rho_{1} = \rho_{0} - AB_{c}Nf_{1}$$

$$\rho_{2} = \rho_{0} - AB_{c}Nf_{1} - AB_{c}Nf_{2}$$

$$\vdots$$

$$\rho_{N} = \rho_{0} - AB_{c}Nf_{1} - AB_{c}Nf_{2} - \dots AB_{c}Nf_{N}$$
(8.4)

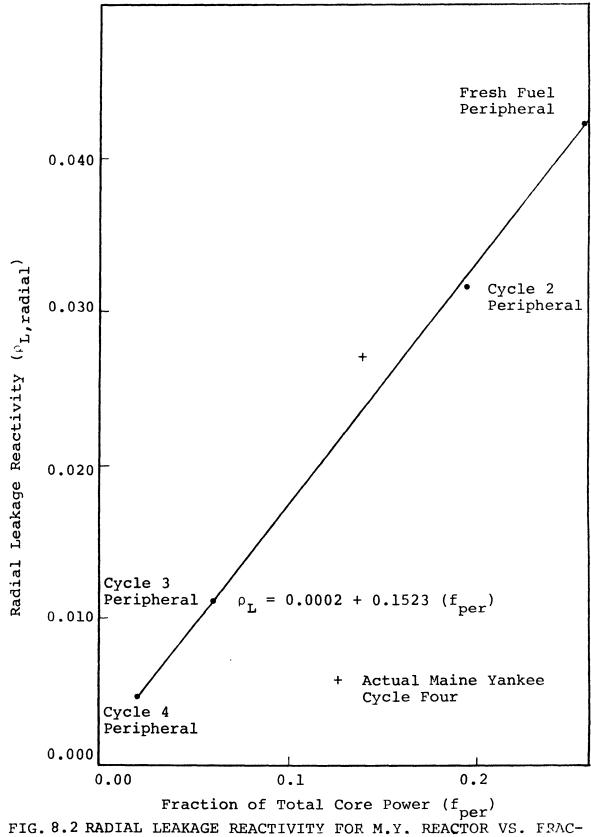
and when the criticality condition is applied:

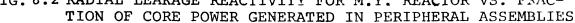
$$B_{\text{DIS}} = \frac{\rho_{0} - \rho_{L}}{A \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i}f_{j} \right\}}$$
(8.5)

The leakage reactivity was found to depend only on the power generated within a few migration lengths of the core periphery. Very good linear correlations were found which permit representation of the leakage reactivity by relations of the form:

$$\rho_{\rm L} = \alpha + \beta f_{\rm per} \tag{8.6}$$

where α and β are constants and f_{per} is the fraction of the total core power generated in the peripheral assemblies (See Fig. 8.2). Hence the discharge burnup becomes:





$$B_{dis} = \frac{\rho_{o} - (\alpha + \beta f_{per})}{A \left(\sum_{i=1}^{N} \sum_{j=i}^{N} f_{i}f_{j} \right)}$$
(8.7)

The cycle schedule index, CSI, given by

$$CSI = \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i}f_{j}$$
(8.8)

quantifies the effects of burnup schedule on uranium utilization, whereas the leakage reactivity, ρ_L , accounts for the leakage effects. If the f_i and the values of f_{per} , α and β are known (for a given assembly type characterized by ρ_0 and A), the uranium utilization can be determined by:

$$U \left(\frac{MWD}{MTF}\right) = \frac{B_{dis} \left(\frac{MWD}{MTP}\right)}{F/P}$$
(8.9)

where F/P is the feed-to-product ratio given by:

$$\frac{F}{P} = \frac{\varepsilon - 0.2}{0.711 - 0.2}$$
(8.10)

for an 0.2 w/o tails assay and reload assembly enrichment $\boldsymbol{\epsilon}$.

The values of α and β depend on the specific reactor design studied, and can easily be extracted from

previous PDQ-7 calculations.

Extensive analytical and numerical studies, too tedious to repeat here, showed that power sharing among batches in a PWR is adequately characterized by the empirical relations

$$f_{i,int} = \frac{\frac{M_{i,int}}{M_{i}} (k_{i})^{\theta}_{int}}{\sum_{\substack{i=1}}^{N} f_{i}}$$
(8.11)

for the interior assemblies of batch i and:

$$f_{i,per} = \frac{\frac{M_{i,per}}{M_{i}} (PNL_{i}k_{i})^{\theta}per}{\sum_{\substack{i=1}}^{N} f_{i}} (8.12)$$

for the peripheral assemblies of batch i, where $M_{i,int}$ is the number of interior assemblies in batch i, $M_{i,per}$ is the number of peripheral assemblies in batch i, $M_{i} = M_{i,int} + M_{i,per}$, k_{i} is the assembly multiplication factor, PNL_i is the non-leakage probability for peripheral assemblies (~ 0.77), and θ_{int} ($\sim 1-2$) and θ_{per} (~ 2) are the values of θ for interior and peripheral asemblies, respectively. The values of θ are design-specific and can be inferred by analysis of representative core maps. Fortunately, fuel discharge burnup values are not particularly sensitive to small changes in θ . The restriction of linear reactivity was relaxed and the foregoing equations were incorporated into the ALARM (A-Linear Advanced Reactivity Model) computer code. Reactivity vs. burnup curves of arbitrary shape can be input to this program; an Nth order Lagrangian interpolation is performed to determine the reactivity of each batch at different burnups. Equations 8.1 -8.12 are extremely versatile and with little or no modifications can be used to self-consistently analyze such design options as low-leakage fuel management, seedblanket cores, variable fuel-to-moderator ratio designs, coastdown, and non-integer batch fractions. Furthermore, the equations can be reformulated for use on axial fuel management problems.

The remainder of the research effort was devoted to application of the preceding methods to the evaluation of a number of suggested improvements in core design and fuel management tactics. The work generally proceeded in three stages: an analytic treatment using the linear reactivity model to determine general trends and identify parameter sensitivity; a numerical evaluation using the ALARM code to obtain quantitative results; and, finally, for certain cases, verification using state-of-the-art physics methods (LEOPARD, PDQ-7).

In the sections which follow, the major areas investigated, and the principal findings in each area, are briefly summarized.

8.4 Axial Blankets*

The use of 6 inch thick blankets of natural uranium or depleted uranium (displacing enriched fuel) were found to provide 2.3% and 2.7% uranium savings, respectively. The use instead of 6 inches of beryllium metal slugs in an axial blanket were found to yield a uranium savings of 1.1%. However, axial power peaking factors are higher than normal for all three cases (by factors of 1.06, 1.075, and 1.053, respectively) and some means (enrichment, poison, or fuel-tomoderator ratio zoning) must be found to reduce these peaking factors before further optimization of axial blanket designs can be pursued.

8.5 Radial Blankets

It was found that assembly-sized natural uranium blankets do not yield any uranium savings for steadystate cores. An N-batch core with a natural uranium blanket can always be surpassed by an N+1 batch core in which the oldest batch is loaded on the core periphery as a "spent fuel" blanket.

^{*} All results using PDQ-7

8.6 Low-Leakage Fuel Management

It was found that low-leakage fuel management schemes for both low and high burnup cores (three batch and five batch fuel management) offer uranium savings of between 2% and 3%. The results are extremely sensitive to the amount of residual burnable poison present in the core. A reduction in this poison residual would increase the savings by an additional 1-2 percent.

8.7 Higher Burnups

Calculations for the Maine Yankee Reactor show that if the discharge burnup could be increased from the current 30,000 MWD/MT to 53,000 MWD/MT with a concurrent shift from 3 batch to 5 batch fuel management, a 15.4% reduction in uranium requirements is possible.

8.8 Optimization of the H/U Atom Ratio

The current Maine Yankee Reactor assembly is near optimum (H/U = 4.38, HOT, BOL) for low burnup operation. For high burnup operation ($\sim 53,000 \frac{MWD}{MTP}$) the assembly should be made a bit wetter: H/U ~ 5.0 (HOT, BOL). This will give a 2.5% reduction in uranium requirements relative to a high burnup, unoptimized (H/U = 4.38) case with the same enrichment.

8.9 Annular Fuel

Annular fuel offers no intrinsic uranium savings per se; savings should be attributed to the concomitant change in the H/U atom ratio and resonance integral. For any annular fuel design a solid pellet design can be found which will have the same uranium utilization. However, annular fuel may provide a convenient way of achieving the uranium savings without significantly altering the assembly design properties.

8.10 Mechanical Spectral Shift

For a current three batch PWR, continuous mechanical spectral shift applied to each batch can provide a theoretical maximum reduction in uranium requirements of 15.5%. Such a scheme, however, may well not be acceptable in light of current thermal/hydraulic and mechanical/operational constraints. Moreover, no one has yet proposed a practical way to continuously vary the H/U ratio in a PWR.

For a mechanical spectral shift scheme of the Concept-1 type (cycle one fuel "dry", cycle two and cycle three fuel "wet") no uranium savings were found and, in fact, uranium losses of a few percent appear to occur, although an all-inclusive optimization over all design space was not performed. For the mechanical spectral shift scheme designated Concept 2 (fresh fuel

starts its first cycle "dry" -- is modified during a short mid-cycle refueling shutdown and finishes the first cycle "wet") it appears that uranium savings on the order of 10% are possible; although, again, an extensive optimization was not performed. Despite these large savings, Concept 2 may be difficult to implement because of thermal/hydraulic considerations, and the decrease in overall capacity factor caused by the additional shutdown required for lattice adjustments. It was also discovered that pin-pulling and bundle reconstitution is <u>mathematically</u> equivalent to coolant void spectral shift (provided the power fractions in the pins are properly calculated).

8.11 Recommendations

Table 8.1 summarizes the potential uranium savings available from design and fuel management changes in current-generation PWRs. The savings should not be considered additive, since only a composite core analysis can determine possible synergisms among the individual changes. This table can serve as the basis for a series of recommendations for follow-on work, for which the need became evident during the course of the investigation which permitted compilation of the subject results.

It is clear from our work and that by other DOE contractors that the use of axial blankets (with an

TABLE 8.1 POTENTIAL URANIUM SAVINGS FROM DESIGN AND FUEL MANAGEMENT CHANGES IN CURRENT PWR CORES

REDUCTION IN CHANGE URANIUM REQUIREMENT COMMENT 0-3%; probably 2% Savings depend on ability to Axial Blankets satisfy axial power peaking constraints via enrichment zoning or burnable poison zoning. Low-Leakage Fuel 2-3%; possibly 5% Savings depend on ability to Management meet radial power peaking constraints with "low-residual" burnable poisons. Reduced assembly width may be useful. Higher Burnup ∿15.4% For increase in enrichment from 3.0 w/o U-235 to 4.34 w/o U-235, and concomitant switch from 3 batch to 5 batch fuel management. Same radial leakage reactivity loss assumed. Natural Uranium Negative Spent fuel is always a better (relative to low leakage blanket material than natural Blankets (Assembly-sized) scheme) uranium.

TABLE 8.1 (Continued)

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CHANGE	REDUCTION IN URANIUM REQUIREMENT	COMMENT
Optimized H/U Atom Ratio	Negligible at low burnup	Although the Maine Yankee reactor is currently opti- mized at a fixed rod dia- meter for H/U atom ratio; other reactors may not be, hence, reductions are reac- tor specific.
	∿2.5% at high burnup	Again, optimization is a reactor design specific result.
Mechanical Spec- tral Shift (Continuous)	<pre>15.5% - three batch (36,000 MWD/MT) 10.3% - five batch (37,000 MWD/MT)</pre>	Impractical in current reactor designs.
Mechanical Spectral Shift (Concept 1: end-of-cycle ad- justment)	Negative (relative to current practice)	Detailed optimization may yet produce some small sav- ings but lattice adjustments are tedious for such small savings.
Mechanical Spectral Shift (Concept 2: mid-cycle adjust- ment)	∿l0% (3 batch core)	Thermal/Hydraulic consider- ations may be limiting. The extra refueling shutdown de- creases the overall reactor capacity factor.

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enrichment less than that of the core fuel) will save uranium provided one can satisfy axial peaking constraints. Further work should be done to pursue this concept to its ultimate embodiment. The models developed in this work provide an indication of the general approach to be followed, namely, reducing peripheral zone power while flattening interior core power. This can be accomplished by one or a combination of the following:

- Enrichment-zoning the core interior, optimizizing the zone lengths and enrichments, and simultaneously optimizing the blanket length and enrichment.
- As above, but using zoned burnable poison to suppress core mid-zone power;
- 3) as above but using a zoned H/U ratio (by varying the diameter of the central void in annular fuel).

The use of beryllium metal as a reflector material should also be investigated further. In addition, multigroup S_N calculations should be performed to verify fast group albedos in all blanket concepts studied, since the use of diffusion theory and one fast group (as in PDQ-7) may not provide a sufficiently accurate description of neutron behavior in the high-leakage core/blanket and blanket/reflector interface zones.

There appears to be a growing interest among utilities in low leakage fuel arrangements in which the radial leakage is reduced by using older fuel assemblies (twice-burned, thrice-burned, etc.) on the core periphery. Peaking factors in the core interior are increased, however, particularly if a concurrent trend to higher burnup is envisioned, requiring increased use of burnable poison to flatten power. This in turn can result in larger poison residuals at EOC. Hence, more work needs to be done to optimize the cycle scheduel index for low leakage fuel management, including the optimum use of burnable poison, evaluation of the use of assemblies which can be split into quarterassemblies and reconstituted (to emulate the BWR with its smaller assemblies), and the use of more than one reload enrichment per reload batch (as in the Russian WWER (VVER), which employs two).

Mechanical spectral shift schemes of the Concept-2 type may provide substantial uranium savings if some practical way could be found to implement them without exceeding thermal-hydraulic safety limits and without decreasing overall reactor capacity factors. One approach which should be evaluated is the incorporation of floodable tubes into the assembly lattice grid.

Finally, it would be of some interest to determine the extent to which the improvments investigated

in this research would also prove beneficial for cores operated in the recycle mode. Low leakage fuel management and the use of axial blankets/axial power shaping would appear directly applicable with comparable savings, for example. APPENDIX A*

DESIGN PARAMETERS FOR MAINE YANKEE REACTOR

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* See Reference S-1. ** See disclaimer in Acknowledgements.

MECHANICAL DESIGN FEATURES OF CYCLE 4 FUEL

	E and F	G, H, and I
Fuel Assembly		
Overall length	156.718**	156.718
Spacer grid size (max. square)		8.115
Retention grid	0	0
No. Zircaloy grids	8	8
No. Inconel grids	1	1
Fuel rod growth clearance	1.021	1.021
Fuel Rod		
Active fuel length	136.7	136.7
Plenum length	8.575	8.575
Clad OD	0.440	0.440
Clad ID	0.384	0.384
Clad wall thickness	0.026	0.028
Pellet OD	0.3765	
Pellet length	0.450	0.450
Dish depth	0.023	0.021
Clad material	Zr-4	
Pellet density initial	95%	94.75%
Initial pressure	(*) psig	(*) psig
Poison Rods		
Overall rod length	146.513	146.322
Clad OD	0.440	0.440
Clad ID	0.388	0.388
Clad wall thickness	0.026	0.026
Pellet OD	0.376	0.376
Clad material	Zr-4	Zr-4

* YAEC 1099P.(Ref. B-6)
** All length dimensions are in inches.

MAIN YANKEE CYCLES 3 AND 4 NUCLEAR CHARACTERISTICS

Core Characteristics	Units	Cycle 3	Cycle 4
Core Average Exposure at BOC	MWD/MT	7,000	10,000
Expected Cycle Length at Full Power	MWD/MT	10,000	9,900
Initial U-235 Enrichment of Fuel Types Type RF 65 Cycle 3 assemblies	w/o	1.93	
Type E 12 Cycle 3, 61 Cycle 4 assemblies	w/o	2.52	2.52
Type F 68 Cycle 3, 12 Cycle 4 assemblies	w/o	2.90	2.90
Type G 32 Cycle 3 and 4 assemblies	w/o	2.73	2.73
Type H 40 Cycle 3 and 4 assemblies	w/o		
Type I 72 Cycle 4 assemblies	W/0		3.03
Control Characteristics			
Number of Control Element Assemblies (CEA's)			
Full Length		77	77
Part Length (not used)		8	*
Total CEA Worth			
HFP, BOC	% ∆p	9.18	8.30
HFP, EOC	¥∆р	9.56	9.30
Burnable Poison Rods			
Number (B_4C in Al ₂ 0 ₃ /Borosilicate glass) Worth at HFP, BOC ²	0 4 -	756/0	
Critical Soluble Boron (ARO) at BOC	€∆р	1.4	0.5
HZP, No Xe, Pk Sm	nnm	1075	1097
HFP, No Xe, Pk Sm	ppm ppm	995	1013
HFP, Equilibrium Xe	ppm	782	797
Reactivity Coefficients (ARO)	rr		121
Moderator Temperature Coefficient			
HFP, BOC	10-4Ap/°F	-0.34**	-0.28
HFP, EOC	10-44p/°F	-0.34** -1.98	-2.31

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Maine Yankee Cycles 3 and 4 Nuclear Characteristics (Cont.)

	Units	Cycle 3	Cycle 4
Fuel Temperature Component of Power Coeff.			
HZP, BOC	10-5Δρ/ ⁰ F	-1.00	-1.53
HFP, BOC	10-540/ ⁰ F	-1.00	-1.18
HZP, EOC	10-5Δρ/ ^O F 10-5Δρ/ ^O F	-1.80	-1.76
HFP, EOC	10-5∆p/ ⁰ F	-1.37	-1.37
Total Delayed Neutron Fraction (β_{eff})			
BOC		0.00611	0.00597
EOC		0.00517	0.00525
Prompt Neutron Generation Time	-6		
BOC	10 ⁻⁶ sec 10 sec	29.3	29.6
EOC	l0 sec	32.3	31.7
Inverse Boron Worth			
HZP, BOC	ppm/%∆ p	84	87
HFP, BOC	ppm/% ∆p	89	93
HZP, EOC	ppm/%∆ p	74	76
HFP, EOC	ppm/%∆ ρ	79	81

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- * Part length CEA's removed for Cycle 4 ** Conditions of 2100 psia.

MAINE YANKEE CYCLE 4 GENERAL SYSTEM PARAMETERS

	Value	Э
Quality	Cycle 3	Cycle 4
Reactor power level (102% of Nominal) (MWT) Average linear heat rate (102% of Nom-	2683	2683
inal)(kw/ft)	6.29	(6.35)*
Peak linear heat generation rate (PLHGR)(kw/ft)	16.5***	(15.7) ¹
Gap conductance at PLHGR (Btu/hr-ft ² - $_{F}^{O}$)	2000***	$(1949)\frac{1}{2}$ (2000) ²
Fuel centerline temp erature at PLHGR (^O F)	3788.1*** 2304.12***	
Fuel average temperature at PLHGR(^O F)	2304.12***	$(2212.9)^{1}$
Hot rod gas pressure (psi)	1221.4***	(1240.5) ¹
<pre>Moderator temperature coefficient at initial density (Δρ/^OF) System flow rate (lbm/hr) Core flow rate (lbm/hr) Initial system pressure (psia) Core inlet temperature (^OF) Core outlet temperature (^OF) Active core height (feet) Fuel rod OD (inches) Number of cold legs Number of hot legs Cold leg diameter (inches) Hot leg diameter (inches) Safety injection tank pressure (psia) Safety injection tank gas/water volume (cu. ft.)</pre>	0.0 134.57x106 130.94x10 2250 554 606.1 11.39 0.440 3 3.3.5 33.5 219.7 2069.7/ 1430.3 683*	0.0 124.57x106 130.94x10 2250 554 606.1 11.39 0.440 3 33.5 33.5 219.7 2069.7/ 1430.3 (3391) ¹
limiting time for PCT	683*	(3391)1 (1385)2

^{*} Parentheses used to denote change in value from previous Cycle 3 reload analysis.

2 For batch I fuel, Cycle 4.

^{**} Cycle 3 used low density fuel from Core IA which was blowdown limited. Cycle 4 does not include any low density fuel so these parameters have been omitted.

^{***} For most limiting high density fuel (batch F) in Cycle 3.

¹ For batch H fuel, Cycle 4.

MAINE YANKEE CYCLE 4

THERMAL HYDRAULIC PARAMETERS AT FULL POWER

General Characteristics		Cycle 3	Cycle 4
Total Heat Output	6 ^{MWT} 10 ⁶ Btu/hr	2630 8976	2630 8976
Fraction of Heat Generated in Fuel Rod Nominal Minimum in Steady State Maximum in Steady State	psig psig	0.975 2235 2185 2285	0.975 2235(2085)† 2185(2035) 2285(2135)
Design Inlet Temperature (steady state) Total Reactor Coolant Flow (design)	0 _F 10 ⁶ 1b/hr	134.5	554(546) 134.6(136.0)
Coolant Flow Through Core (design)	10 ⁶ 1b/hr	130.7	130.7(132.1)
Hydraulic Diameter (nominal channel)	6 2	0.044	0.044
Average Mass Velocity Pressure Drop Across Cross	10^{6} lb/hr-ft ²	2.444	2.444(2.47)
(design flow) Total Pressure Drop Across	psi	9.7	9.7(9.9)
Vessel (based on nominal dimensions & design flow) Core Average Heat Flux Total Heat Transfer Area Film Coofficient at Average	psi Btu/hr-ft ² ft ²		32.4(33.1) 180,575* 48,480*
Film Coefficient at Average Conditions Maximum Clad Surface Temp.	Btu/hr-ft ² - ⁰ F	5640 656	5636 656
Average Film Temperature Dif- ference Avg. Linear Heat Rate of Rod Average Core Enthalpy Rise	o _F kw∕ft Btu∕lb	31.7 6.03*	32 6.09* 68.7
Calculational Factors Engineering Heat Flux Factor		1.03	1.03
Engineering Factor on Hot Channel Heat Input Flow Factors		1.03	1.07
Inlet Plenum Nonuniform Dis	_	1.05	1.05
tribution End Pitch, Bowing and Clad Diameter		1.065	1.065

* Allows 0.3 percent axial shrinkage due to fuel densification.

 Numbers in parentheses are provided for conditions at a nominal pressure of 2,085 psig and a design inlet temperature of 546°F.

APPENDIX B

SPECIAL USES AND MODIFICATIONS OF THE LEOPARD PROGRAM

At end-of-cycle most PWRs have essentially no soluble boron remaining in the coolant. Hence, LEOPARD results corresponding to the EOC burnup of each fuel batch should have no boron present. However, because the power fractions in each batch are different, the average burnup logged by each fuel batch (newest through oldest) will be different, and, hence, the correct boron coastdown endpoints are difficult to estimate.

The presence of boron in the fuel assembly forces an increase in the fast/thermal flux ratio to keep the thermal power output constant. Hence, the fast flux is increased and more Pu-239 is produced. It was found, however, that the use of the cycle-average boron concentration was adequate to match the isotopic concentrations produced by using the more exact boron coastdown treatment. In essence, the correct "average" spectrum is used. In order to obtain zero boron EOC k_{∞} estimates, however, the following methodology was developed:

 LEOPARD supercell calculations are run with the correct <u>cycle-average</u> boron concentration and the usual depletion time step sizes.

2) After every depletion time step (usually about 2500 MWD/MT) the boron concentration is set to zero and a very short (<1 MWD/MT) time step is used to find the k_{∞} value

with no boron present. (If it is desirable to re-equilibrate xenon, then these time steps should be ~24 MWD/MT; this effect was found to be negligible in the present instance).

3) The cycle average boron is restored to the supercell and the next long depletion time step is run.

4) This procedure is continued until the projected discharge burnup is reached (or exceeded).

In this manner, the k_{∞} values with no boron present can be determined and used for the EOC criticality calculation, but they will have been produced with the correct (boron present) spectrum.

For high burnup cases, a LEOPARD supercell which has been computed with no boron in the coolant can have a discharge burnup as much as 3% lower than the same supercell run with the method just described.

Note, however, that the key to estimating relative uranium savings is to compare cases on a consistent basis. Hence, if two cases are run without boron and the same two cases are run using the above boron removal technique, the relative advantage of one case over the other will be the same.

The LEOPARD code contains the option to vary the D_2O/H_2O molecule ratio as a function of burnup. For example, an input of -1.000 on the 101 poison control card gives a lattice with 50% H_2O and 50% D_2O . In order to input voids as a percentage of coolant volume (void fraction in coolant), the following modifications were made in two LEOPARD subroutines:

In subroutine INPUT change:

	<pre>PUREN(31,L)=WATER/(1.0+DENIMIC(I))</pre>	213
	PUREN(2,L)=PUREN(2,L)+PUREN(31,L)+PUREN(32,L)+2.*(PUREN(28.L)+	227
1	PUREN(29,L)+PUREN(30,L))	228
	PUREN(20,L)=PUREN(20,L)+2.*PUREN(32,L)	243

to:

915

915

	<pre>PUREN(31,L)=WATER*(1.0-DENIMIC(I))</pre>	213
	PUREN(2,L)=PUREN(2,L)+PUREN(31,L)+2.*(PUREN(28,L)+	227
1	PUREN (29,L) + PUREN (30,L)	228
	PUREN(20,L)=PUREN(20,L)	243

and in subroutine BURN change:

135	<pre>PUREN(31,J)=WATER/1.0-POISON(NOSTEP+1))</pre>	155
	PUREN $(20, J) = 2.0 * PUREN(32, J)$	158

to:

Hence, entering -0.5 on the 101 poison entry card will give a lattice with 50% voids and 50% H_2O . The poison values or the depletion timestep cards are similarly specified.

APPENDIX C

OPTIMIZATION OF CYCLE SCHEDULE INDEX

The general equation for the discharge burnup of an N-batch core is:

$$B_{dis} = \frac{\rho_{o} - \rho_{L}}{A \left\{ \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j} \right\}}$$
(C.1)

where ρ_0 is the initial extrapolated reactivity, ρ_L is the leakage reactivity, A is the slope of the reactivity vs. burnup curve, and f_i is the power fraction in batch i. Selection of a set of power fractions to produce the maximum discharge burnup involves the cycle schedule index and the leakage reactivity, ρ_L , since the leakage reactivity can be written as:

$$\rho_{\rm L} = \alpha + \beta f_{\rm per} \tag{C.2}$$

It is useful, however, to consider the zero-leakage limit, which requires minimization of the cycle schedule index, CSI. The equation for the CSI is:

$$CSI = \sum_{i=1}^{N} \sum_{j=i}^{N} f_{i} f_{j}$$
(C.3)

subject to the constraint

$$\sum_{i=1}^{N} f_i = 1.0$$
 (C.4)

The equation for CSI really represents an equation for

ellipses (or elliptical solids depending on the value of N). In two dimensions (N=2) the CSI is written

$$CSI = f_1^2 + f_1 f_2 + f_2^2$$
 (C.5)

In a graphical representation, different values of the CSI trace out concentric ellipses having their minor radius along a line at a 45° angle to the f_1 axis. Since the CSI is proportional to the length of the minor radius, minimization of the CSI means finding the smallest minor radius which will satisfy the constraint equation (see Fig. C.1). This occurs when the ellipse is first tangent to the constraint line, or $f_1 = f_2$.

The result can be generalized to higher dimensions (N>2). For instance, for N=3 the CSI describes an elliptical solid, and the constraint equation $(f_1 + f_2 + f_3 = 1)$ describes a plane. The smallest elliptical solid (minimum CSI) which will satisfy the constraint is the elliptical solid just tangent to the surface, hence $f_1 = f_2 = f_3$. Hence in the absence of peripheral leakage a flat power profile/uniform power history maximizes reactivity limited burnup.

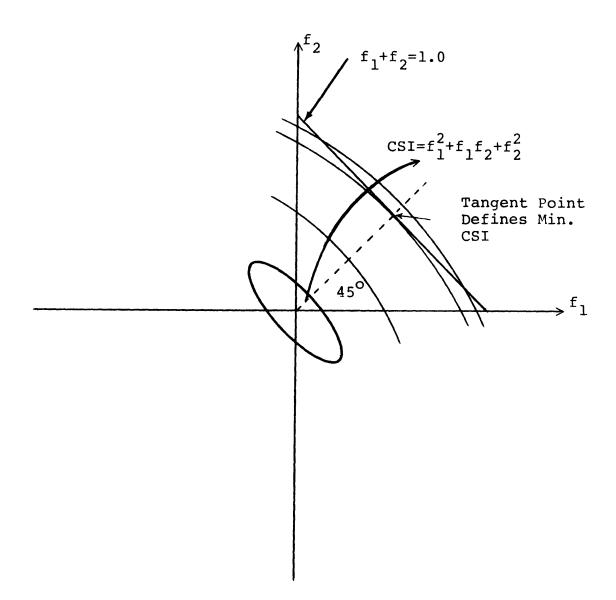


FIG. C.1 BEHAVIOR OF CSI RELATIVE TO THE CONSTRAINT EQUATION $(f_1+f_2=1.0)$ FOR N=2.

APPENDIX D

DEVELOPMENT OF THE 0 FORMULATION

Consider the two-bundle, zero-current boundary condition problem of Fig. 3.4. Using group-and-one-half theory it is possible to write the power in bundle one as

$$q_1 = c_1 k_1 \phi_{11} \tag{A.1}$$

where q_1 is the bundle power (relative to that of the average bundle), C_1 is a constant, k_1 is the infinite medium multiplication factor of region one, and ϕ_{11} is the fast group flux in region one. We can, by fiat, lump $C_1\phi_{11}$ into one term and say:

$$C_1 \phi_{11} = k_1^{N-1}$$
 (A.2)

This is a definition, not an assumption; and hence:

$$q_1 = k_1^{N}$$
(A.3)

The same thing can be done for bundle two to give

$$q_2 = k_2^{M}$$
(A.4)

For the zero-current boundary condition problem

$$\frac{1}{k_{\rm sys}} = \frac{f_1}{k_1} + \frac{f_2}{k_2}$$
(A.5)

where:

$$f_1 = \frac{q_1}{q_1 + q_2}$$
; $f_2 = \frac{q_2}{q_1 + q_2}$ (A.6)

Thus, combining terms gives

$$k_{sys} = \frac{k_1^{N} + k_2^{M}}{k_1^{N-1} + k_2^{M-1}}$$
(A.7)

We can arbitrarily define:

$$N = \theta + \Delta N$$

$$M = \theta + \Delta M$$
(A.8)

where $\boldsymbol{\theta}$ is an unknown, to give:

$$k_{sys} = \frac{k_1^{\theta} k_1^{\Delta N} + k_2^{\theta} k_2^{\Delta M}}{k_1^{\theta-1} k_1^{\Delta N} + k_2^{\theta-1} k_2^{\Delta M}}$$
(A.9)

But it is always possible to find a ΔN and a ΔM such that

$$\frac{k_1^{\Delta N}}{k_2^{\Delta M}} = 1.0$$
 (A.10)

Thus, multiplying the numerator and denominator of Eq. A.9 by $k_2^{\ \Delta M}$ gives:

$$k_{sys} = \frac{k_1^{\theta} + k_2^{\theta}}{k_1^{\theta-1} + k_2^{\theta-1}}$$
(A.11)

It is important to recognize that no inherent assumptions are contained in this derivation--only definitions are used, i.e., $q_1 = k_1^N$ is completely arbitrary since given a q_1 and a k_1 , N is defined. The importance of this formula (Eq. A.11) for system reactivity and power split calculations lies in the fortunate result (see Fig. 4.2) that θ does not change significantly for a wide range of k_1 and k_2 combinations produced by a large variety of lattice adjustments. The value of theta does depend on problem size (i.e. the width of the separate regions (k_1, k_2) inside the zero-current boundary condition perimeter.

ALARM CODE INPUT SPECIFICATIONS, LISTING, AND SAMPLE PROBLEM

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APPENDIX E

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E.l Code Description

The ALARM (<u>A-Linear Advanced Reactivity Model</u>) code is the equivalent of a poor man's PDQ-7. It enables one to calculate the average discharge burnup of a fuel batch including batch power sharing and radial and axial leakage effects. These effects are calculated using the methodology shown in Table 4.3 and Fig. 4.9. The format for input preparation is shown in Table E.1. The code is written in FORTRAN (Table E.2) but can easily be adapted to BASIC for use with minicomputers.

The sample problem uses LEOPARD results for a 4.34 w/o U-235 Maine Yankee (C-E design) assembly in a C-E SYSTEM 80^{TM} core. The leakage correlation is based on the "near-equilibrium" data points of Fig. 3.8 with a reactivity decrement of 0.005Ap added for the addition of burnable poison. A lowleakage loading pattern is used based on the data in Ref. M-3. The card images for the input are shown in Table E.3. Table E.4 is a listing of the code output.

TABLE E.1 INPUT INSTRUCTIONS

Card Set 1 (one card)

J: number of batches

IDEG: order of Lagrangian interpolation

TSTEP: time-step size (MWD/MT)

CONVRG: convergence criteria (MWD/MT)

FORMAT (8X, 12, 8X, 12, 2F10.4)

Card Set 2 (one or more cards)

PNL(I): batch non-leakage probabilities; J entries

FORMAT (8F10.3)

Card Set 3 (one or more cards)

AINT(I): number of internal assemblies in batch I; J entries FORMAT (8F10.1)

Card Set 4 (one or more cards)

APER(I): number of peripheral assemblies in batch I; Jentries FORMAT (8F10.1)

Card Set 5 (one card)

ALPHA BETA leakage coefficients for $\rho_{L} = \alpha + \beta f_{per}$ FORMAT (2F10.4)

<u>Card Set 6</u> (one card) THETAI: interior assembly θ value THETAP: peripheral assembly θ value FORMAT (2F10.2)

Card Set 7 (one card) N: number of (k, BURNUP) pairs entered FORMAT (2X, I3)

Card Set 8 (one or more cards)

X(I): burnup values (MWD/MT); N entries
FORMAT (8F10.4)

<u>Card Set 9</u> (one or more cards) Y(I): value of k_{∞} at each burnup; N entries FORMAT (8F10.4)

Card Set 10

Termination Card: enter 99; FORMAT (2X, I3)

TABLE E.2 ALARM CODE LISTING

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.

0001		IMPLICIT REAL+8(A-H, O-Z) , REAL+8(K)
0002		DIMENSION X(100), Y(100)
0003		DIMENSION DONE(20), B(20), F(20), K(20), PNL(20)
0004		DIMENSION AINT(20), APER(20), PINT(20), PPER(20), ATOT(20)
	С	
0005	20	READ(5,1020) J,IDEG,TSTEP,CONVRG
0006		READ (5,1040) (PNL(I), I=1,J)
0007		READ (5,1050) (AINT(I), I=1,J)
0008		READ (5,1050) (APER(I), I=1,J)
0009		READ (5,1060) ALPHA, BETA
0010		READ (5,1030) THETAI,THETAP
	С	,
0011		STORE = TSTEP
	С	READ N XANDY VALUES AND PRINT
0012	11	READ(5,1000) N
0013		IF (N.EQ.99) STOP
0014		READ(5,1010)(X(I),I=1.N)
0015		READ (5,1010) (Y(I), I=1.N)
0016		TSTEP = STORE
0017		NUMB = 0
0018		ATOTAL = 0.0
0019		DO 122 I=1,J
0020		ATOTAL = ATOTAL + APER(I) + AINT(I)
0021		ATOT(I) = APER(I) + AINT(I)
0022	122	CONTINUE
	С	
0023		WRITE(6,3000)
0024		WRITE(6,3010) J,ATOTAL, IDEG, TSTEP, CONVRG, THETAI, THETAP, ALPHA, BETA
0025		WRITE(6,3020)
0026 .		DO 10 I = 1, N
0027	10	WRITE(6,3030) I,X(I),Y(I)
0028		WRITE(6,3040)
0029		DO 15 I=1,J
0030	15	WRITE(6,3050) I,AINT(I),APER(I),ATOT(I),PNL(I)
•	C	
0031		WRITE(6,3060)

c	,	
	S=1.0	
	RHOL=ALPHA+BETA+0.2	
	V=J	
	UONE(1)=0.0	
	DO 100 I=1, N	
	GAMMA=1.0+RHOL	
	IF(Y(I).LE.GAMMA.AND.X(I).GT.10000.0) BEST=X(I)+2/(V+1)	
	IF(Y(I).LE.GAMMA.AND.X(I).GT.10000.0) GD TO 110	
	CONTINUE	
0042 110	CONTINUE	
	DD 120 I=1,J	
0046	DONE(I)=W+BEST	
	CONTINUE	
0048 C 130	N N	
	F(I) = (APER(I) + AINT(I)) / ATOTAL	
0050 140 C	CONTINUE	
-		
	DO 200 I=1, d R(I)= DONE(I)	
	B(1)=DONE(1)	
	CONTINUE	
	GO TO 330	
C C		
	CONTINUE	
	NUMB = NUMB + 1	
	WRITE(6,3070)NUMB ,RHOL	
	WRITE(6,3080) FPER	
	WRITE(6,3090) (B(I),I=1,J)	
	WRITE (6,3100) (F(I),I=1,J)	
	IF(DABS(TELL).LE.CONVRG) GO TO 11	
C C		
	TSTEP = STORE	
-	S=0.0	
	DO 230 I=1,J	
	L=I-1	
· · · = =	IF (I.NE.1) DONE(I)=B(L)	
	CONTINUE	
	GO TO 190	
C		
6		

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0069	330 DO 370 I=1,J
0070	XARG=B(I)
	C THIS SECTION WILL LOCATE XARG RELATIVE TO THE TABLE VALUES
	C INPUT AND CENTER THE INTERPOLATION POLYNOMIAL ABOUT XARG
	C IT CAN ALSO SET UP THE POLYNOMIAL FOR EXTERPOLATION
0071	340 DD 350 L=1,N
0072	IF (L.EQ.N.DR.XARG.LE.X(L)) GO TO 360
0073	350 CONTINUE
0074	360 MAX = L + IDEG/2
0075	IF (MAX.LE.IDEG) MAX = IDEG+1
0076	IF (MAX.GE.N) MAX=N
0077	MIN = MAX-IDEG
	C THE VALUE OF MIN HAS NOW BEEN DETERMINED FOR FLAGR TO USE
	c .
	C
0078	YINTER = FLAGR (X,Y,XARG,IDEG,MIN,N)
0079	K(I)=YINTER
0080	370 CONTINUE
	C
	C · · ·
	C C
0081	C KINV=0.0
0081	C
0081	C KINV=0.0
0081 0082	C KINV=0.0 380 DO 390 I=1,J
	C KINV=0.0
0082	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE
0082 0083	C KINV=0.0 380 DD 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHD=1-KINV
0082 0083 0084	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHD=1-KINV IF (TSTEP.GT.0.1) GD TD 400
0082 0083 0084 0085	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHD=1-KINV IF (TSTEP.GT.0.1) GD TD 400 IF (RHD.LE.RHOL) GD TD 460
0082 0083 0084 0085 0086	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHD=1-KINV IF (TSTEP.GT.0.1) GD TD 400
0082 0083 0084 0085 0086 0087	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHO=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RHO.LE.RHOL) GO TO 460 400 IF (RHO.GT.RHOL) GO TO 440 410 DO 420 I=1,J
0082 0083 0084 0085 0086 0087 0088	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHO=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RHO.LE.RHOL) GO TO 460 400 IF (RHO.GT.RHOL) GO TO 440
0082 0083 0084 0085 0086 0087 0088 0088	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHO=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RHO.LE.RHOL) GO TO 460 400 IF (RHO.GT.RHOL) GO TO 440 410 DO 420 I=1,J
0082 0083 0084 0085 0086 0087 0088 0089 0089	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RHO=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RHO.LE.RHOL) GO TO 460 400 IF (RHO.GT.RHOL) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))+F(I)+TSTEP)
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))+F(I)+TSTEP) 420 CONTINUE
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))*F(I)*TSTEP) 420 CONTINUE TSTEP=TSTEP/2
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))*F(I)*TSTEP) 420 CONTINUE TSTEP=TSTEP/2 C C C 440 DD 450 I=1,J
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091 0092	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))*F(I)*TSTEP) 420 CONTINUE TSTEP=TSTEP/2 C C C 440 DD 450 I=1,J
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091 0092	C KINV=0.0 380 DO 390 I=1,J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1,J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))+F(I)+TSTEP) 420 CONTINUE TSTEP=TSTEP/2 C C
0082 0083 0084 0085 0086 0087 0088 0089 0090 0091 0092	C KINV=0.0 380 DO 390 I=1.J KINV=KINV+(F(I)/K(I)) 390 CONTINUE RH0=1-KINV IF (TSTEP.GT.0.1) GO TO 400 IF (RH0.LE.RH0L) GO TO 460 400 IF (RH0.GT.RH0L) GO TO 440 410 DO 420 I=1.J B(I)=B(I)-((ATOTAL/(APER(I)+AINT(I)))+F(I)+TSTEP) 420 CONTINUE TSTEP=TSTEP/2 C C 440 DO 450 I=1.J B(I)=B(I)+((ATOTAL/(APER(I)+AINT(I)))+F(I)+TSTEP)

	c	
0096		PTOTAL = 0.0
0097	470	DO 480 I=1.J
0098	476	PTOTAL=PTOTAL+APER(I)+(PNL(I)+K(I))++THETAP +
0030	s	AINT(I)+K(I)++THETAI
0099	-	CONTINUE
	C	
	č	
0100	- 490	DO 500 I=1,J
0101		<pre>PPER(I)=(APER(I)+(PNL(I)+K(I))++THETAP)/PTOTAL</pre>
0102		PINT(I)=(AINT(I)+K(I)++THETAI)/PTOTAL
0103		F(I) = PPER(I) + PINT(I)
0104	500	CONTINUE
0105		FPER=0.0
0106	510	DO 520 I=1,J
0107		FPER=FPER+PPER(I)
0108	520	CONTINUE
0109		RHOL= ALPHA+ BETA+ FPER
0110		GD TD 330
0111		CONTINUE
0112		IF (J.EQ.1) GO TO 560
0113	550	DD 560 I=1,JLESS
0114		M=I+1
0115		TELL=B(I)-DONE(M)
0116		IF (DABS(TELL).GT.CONVRG) GO TO 210
0117		IF (S.EQ.1) GO TO 210
0118	560	CONTINUE
0119		GO TO 210 -
0120		STOP
	С	
	C	FORMATS FOR INPUT AND OUTPUT STATEMENTS
0121		FORMAT (2X, I3)
0122		FORMAT (8F10.4)
0123		FORMAT (8X, 12, 8X, 12, 2F10.4)
0124		FORMAT(2F10.2)
0125	1040	FORMAT (8F10.3)

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0126	1050 FORMAT (8F10.1)
0127	1060 FORMAT (2F10.4)
0128	3000 FORMAT(1H1.9X, THE INPUT PARAMETERS ARE : ///1H)
0129	3010 FORMAT (5X, 1HJ, 10X, 17HNUMBER OF BATCHES, 16X, 15//
0.29	\$ 5X,6HATOTAL,5X,26HTOTAL NUMBER OF ASSENBLIES,7X,F5.1//
	\$ 5X,4HIDEG,7X,19HORDER OF REGRESSION,17X,12//
	\$ 5X.5HTSTEP.6X.18HTIME STEP (MWD/MT).15X.F5.1//
	\$ 5X,6HCONVRG,5X,30HCONVERGENCE CRITERION (MWD/MT),3X,F5.1//
	\$ 5X,6HTHETAI,5X,14HINTERNAL THETA,19X,F5.1//
	\$ 5X,6HTHETAF,5X,16HPERIPHERAL THETA,17X,F5.1//
	\$ 5X.5HALPHA.6X.19HLEAKAGE COEFFICIENT.12X.F7.4//
	\$ 5X,4HBETA,7X,19HLEAKAGE COEFFICIENT,13X,F6.4/1H1)
0130	3020 FORMAT(10X, 'THE READ IN VALUES OF K-EFFECTIVE VS. BURNUP'///20X,
0.50	\$ 14HBURNUP(MWD/MT), 15X, 5HK-EFF/20X, 14(1H+), 14X, 7(1H+)/1H0)
0131	3030 FORMAT(8X, I2, 11X, F10.2, 16X, F7.4)
0132	3040 FORMAT (1H1,4X, 'NON-LEAKAGE PROBABILITY AND THE DISTRIBUTION OF
	\$ASSEMBLIES IN THE BATCHES'///20X,8HINTERNAL,7X.
	\$ 10HPERIPHERAL, 5X, 5HTOTAL, 1X, 5HBATCH, 4X, 11HNON-LEAKAGE/
	\$ 10X, 5HBATCH, 3(5X, 10HASSEMBLIES), 5X, 11HPROBABILITY/
	\$ 10X, 5(1H*), 3(5X, 10(1H*)), 5X, 11(1H*)/1H0)
0133	3050 FORMAT(11X, 12,9X,F5.1,2(10X,F5.2),12X,F5.3/1H0)
0134	3060 FORMAT(1H1.4X' THE VALUES OF LEAKAGE (RHOL), PERIPHERAL BATCH
	\$POWER FRACTION (FPER).'/ 5X,' BATCH BURNUP B(I) IN MWD/MT AND
	S BATCH POWER FRACTION F(I) ARE ARRANGED'/ 5X,' IN THE FOLLOWING
·	\$ ORDER AFTER EACH ITERATION'///13X,4HRHOL/13X,4HFPER//13X,4HB(1),
	\$ 9X,4HB(2), 9X,4HB(3), 9X,4HB(4), 9X,4HB(5), 9X,4HB(6).
	\$ 6X,3HETC/13X,4HF(1), 9X,4HF(2), 9X,4HF(3), 9X,4HF(4),
	\$ 9X,4HF(5), 9X,4HF(6),6X,3HETC///)
0135	3070 FORMAT(/3X, I3, 2X, F10.4)
0136	3080 FORMAT (8X.F10.4)
0137	3090 FORMAT (5X.8(3X.F10.1))
0138	3100 FORMAT (5X.8(3X.F10.4)/1H)
	c c
	END

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0001	FUNCTION FLAGR (X,Y,XARG,IDEG,MIN,N) * C	
	C FLAGR USES THE LAGRANGE FORMULA TO EVALUATE THE INTERPOLATING	
	C POLYNOMIAL OF DEGREE IDEG FOR ARGUMENT XARG USING THE DATA	
	C VALUES X (MIN)X (MAX) AND Y (MIN)Y (MAX) WHERE	
	C MAX = MIN + IDEG. NO ASSUMPTION IS MADE REGARDING ORDER OF	
	C THE X(I) AND NO ARGUMENT CHECKING IS DONE. TERM IS	
	C A VARIABLE WHICH CONTAINS SUCCESSIVELY EACH TERM OF THE C Lagrange 'Formula. The Final value of yest is the interpolated	
	C LAGRANGE FORMULA. THE FINAL VALUE OF YEST IS THE INTERPOLATED C VALUE.	
0002	IMPLICIT REAL+8(A-H, O-Z)	
0003	REAL*B X,Y, XARG, FLAGR	
0004	DIMENSION X (N), Y (N)	
	C	
0005	FACTOR = 1.0	
0006	MAX = MIN + IDEG	
0007	DO 3002 J=MIN,MAX	
0008	IF (XARG.NE.X(J)) GO TO 3002	
0009	FLAGR = Y(J)	
0010	RETURN	
0011	3002 FACTOR = FACTOR+(XARG - X(J))	
	C EVALUATE THE INTERPOLATING POLYNOMIAL	
0012	C EVALUATE THE INTERPOLATING POLYNOMIAL YEST = 0.	
0013	DO 3005 I=MIN.MAX	
0014	TERM = Y(I) * FACTOR/(XARG - X(I))	
0015	DD 3004 J=MIN.MAX	
0016	3004 IF (I.NE.J) TERM = TERM/(X(I)-X(J))	
0017	3005 YEST = YEST + TERM	
0018	FLAGR = YEST	
0019	RETURN	
	C	
0020	END	

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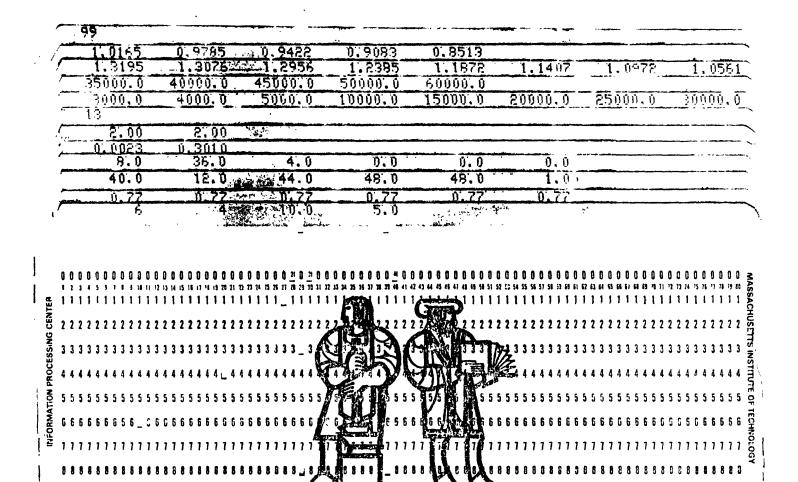
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* REF. (C-7)

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THE INPUT PARAMETERS ARE :

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J	NUMBER OF BATCHES	6
ATOTAL	TOTAL NUMBER OF ASSENBLIES	241.0
IDEG	ORDER OF REGRESSION	4
TSTEP	TIME STEP (MWD/MT)	10.0
CONVRG	CONVERGENCE CRITERION (MWD/MT)	5.0
THETAI	INTERNAL THETA	2.0
THETAP	PERIPHERAL THETA	2.0
ALPHA	LEAKAGE COEFFICIENT	0.0023
BETA	LEAKAGE COEFFICIENT	0.3010

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THE READ IN VALUES OF K-EFFECTIVE VS. BURNUP

	BURNUP(MWD/MT)	K-EFF
	*********	******
	•	
1	3000.00	1.3195
2	4000.00	1.3076
3	5000.00	1.2956
4	10000.00	1.2385
5	15000.00	1.1872
6	20000.00	1.1407
7	25000.00	1.0972
8	30000.00	1.0561
9	35000.00	1.0165
10	40000.00	0.9785
11	45000.00	0.9422
12	50000.00	0.9083
13	60000.00	0.8513

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NON-LEAKAGE PROBABILITY AND THE DISTRIBUTION OF ASSEMBLIES IN THE BATCHES

BAT CH *****	INTERNAL ASSEMBLIES	PERIPHERAL ASSEMBLIES *********	TOTAL BATCH Assemblies ********	NON-LEAXAGE PROBABILITY ********
1	40.0	8.00	48.00	0.770
2	12.0	36.00	48.00	0.770
3	44.0	4.00	48.00	0.770
4	48.0	0.0	48.00	0.770
5	48.0	0.0	48.00	0.770
6	1.0	0.0	1.00	0.770

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THE VALUES OF LEAKAGE (RHOL), PERIPHERAL BATCH POWER FRACTION (FPER), BATCH BURNUP B(I) IN MWD/MT AND BATCH POWER FRACTION F(I) ARE ARRANGED IN THE FOLLOWING ORDER AFTER EACH ITERATION

	RHOL FPER						
. ·	B(1) F(1)	B(2) F(2)	B(3) F(3)	B(4) F(4)	B(5) F(5)	B(6) F(6)	ETC ETC
1	0.0474 0.1499						
•	18095.4 0.2475	20835.7	31746.6 0.2077	39097.5 0.1922	46170.6 0.1727	53423.7 0.0032	
2	0.0455 0.1435						
	14030.1 0.2659	26236.4 0.1634	31475.9 0.2098	41158.9 0.1874	47577.2 0.1702	53858.3 0.0032	
3	0.0474 0.1500						
	13077.5 0.2695	22055.9 0.1739	35394.4 0.1971	40252.1 0.1895	48804.3 0.1668	54575.1 0.0032	
4	0.0478						
	13524.5 0.2674	21511.9 0.1753	32120.4 0.2072	43991.8 0.1790	48276.0 0.1680	55941.3 0.0031	

5	0.0476					
	13635.0	21974.7	31741.3	41205.2	51673.8	55525.5
	0.2671	0.1742	0.2085	0.1869	0.1602	0.0032
	0.2071	0.1/42	V.2005	4.1009		0.0032
6	0.0473				•	
	0.1495					
	14011.9	22316.9	32429.9	41140.8	49424.9	58821.7
	0.2653	0.1732	0.2063	0.1869	0.1653	0.0030
7	0.0473					
	0.1494		•			
	13757.9	22480.4	32525.2	41560.8	49210.1	56635.5
	0.2665	0.1728	0.2060	0.1858	0.1658	0.0031
-						
8	0.0474					
	0.1499					
	13670.8	22201.9	32597.8	41583.0	49528.9	56391.8
	0.2668	0.1735	0.2058	0.1857	0.1650	0.0031
9	0.0474					
-	0.1499	•				
	13728.1	22163.1	32403.7	41686.1	49584.0	56713.2
	0.2666	0.1736	0.2064	0.1854	0.1649	0.0031
	-					
10	0.0474					
	0.1498					
	13751.0	22228.1	32388.3	41533.0	49689.7	56776.0
	0.2665	0.1734	0.2064	0.1859	0.1647	0.0031
11	.0474					
	J.1498					
	13760.1	22253.9	32450.8	41526.0	49559.4	56877.0
	0.2665	0.1734	0.2062	0.1859	49559.4	0.0031
	4.5003	V11704	V. 2VU2	V11033	A. 103A	4.0431

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12	0.0474					
	0.1498					
	13741.9	22250.0	32458.8	41567.7	49541.9	56748.5
	0.2665	0.1734	0.2062	0.1858	0.1650	0.0031
13	0.0474					
	0.1498					
	13737.2	22231.0	32451.8	41571.4	49576.0	56730.0
	0.2666	0.1734	. 0.2062	0.1858	0.1649	0.0031
14	0.0474					
	0.1498					
	13742.9	22230.7	32440.0	41569.3	49582.8	56764.2
	0.2665	0.1734	0.2063	0.1858	0.1649	0.0031
15	0.0474					
15						
	0.1498					
	13744.8	22236.9	32441.2	41560.3	49582.1	56771.4
	0.2665	0.1734	0.2063	0.1858	0.1649	0.0031
16	0.0474					
	0.1498					
	13744.3	22238.2	32446.1	41561.0	49573.9	56770.5
	0.2665	0.1734	0.2062	0.1858	0.1649	0.0031
	0.2000	0.1754	0.2002	0.1000	0.1049	0.0001
17	0.0474					
	0.1498					
	13743.0	22237.0	32446.3	41564.4	49573.6	56762.3
	0.2665	0.1734	0.2062	0.1858	0.1649	0.0031
	4.2000	0.1704		0.1000	011049	0.0031

APPENDIX F

CYCLE SCHEDULE INDICES FOR NON-INTEGER BATCH NUMBERS

Consider a reactor with total number of assemblies N_{tot} , of which $3N_A$ assemblies are run in a three batch mode and $2N_B$ assemblies are run in a two batch mode. Hence, at every shutdown N_A assemblies of type A are loaded and discharged and N_B assemblies of type B are loaded and discharged. The governing reactivity equations can be written as:

$$\rho_{A1} = \rho_{OA} - A_{A} B_{C} \frac{f_{1}}{(N_{A}/N_{tot})}$$

$$\rho_{A2} = \rho_{OA} - A_{A} B_{C} \frac{(f_{1} + f_{2})}{(N_{A}/N_{tot})}$$

$$\rho_{A3} = \rho_{OA} - A_{A} B_{C} \frac{(f_{1}^{+}f_{2}^{+}f_{3})}{(N_{A}/N_{tot})}$$

$$\rho_{B1} = \rho_{OB} - A_{B} B_{C} \frac{f_{4}}{(N_{B}/N_{tot})}$$

$$\rho_{B2} = \rho_{OB} - A_{B} B_{C} \frac{(f_{4}^{+} + f_{5})}{(N_{B}/N_{tot})}$$
(F.1)

where ρ_{Ai} (and ρ_{Bi}) are the reactivities of sublots A (and B) during cycle i of in-core residence, ρ_{OA} (and ρ_{OB}) are the initial reactivities of fuel types A (and B), ${}^{A}_{A}$ (and ${}^{A}_{B}$) are the slopes of the reactivity vs. burnup curves for fuel types A (and B), and B_C represents the core average cycle burnup (MWD/MTP). The quantity $f_{1}/(N_{A}/N_{tot})$ can be interpreted as the relative energy generated in batch 1 of sublot A in an equilibrium burnup cycle; and similarly for the other analogous terms. Application of the criticality condition:

$$\rho_{\mathbf{L}} = \sum_{i=1}^{N} \mathbf{f}_{i} \rho_{i}$$
 (F.2)

gives:

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$$B_{C} = \frac{(f_{1}+f_{2}+f_{3}) \rho_{OA}+(f_{4}+f_{5})\rho_{OB}-\rho_{L}}{A_{A}\left(\frac{N_{tot}}{N_{A}}\sum_{i=1}^{3}\int_{j=i}^{3}f_{i}f_{j}\right) + \left(\frac{N_{tot}}{N_{B}}\sum_{i=4}^{5}\int_{j=i}^{5}f_{i}f_{j}\right) A_{B}}$$
(F.3)

If we assume that type A and type B fuel are identical then:

$$B_{C} = \frac{\rho_{o} - \rho_{L}}{A \left[\frac{N_{tot}}{N_{A}} \left[\sum_{i=1}^{j} \sum_{j=i}^{j} f_{i}f_{j} \right] + \frac{N_{tot}}{N_{B}} \left[\sum_{i=4}^{j} \sum_{j=i}^{j} f_{i}f_{j} \right] \right]}$$
(F.4)

The discharge burnup of type A fuel (three residence cycles) is given by:

$$B_{dis,A} = B_{c} \frac{N_{tot}}{N_{A}} (f_{1} + f_{2} + f_{3})$$
 (F.5)

and the discharge burnup for type B fuel (two residence cycles) is given by

$$B_{dis,B} = B_C \frac{N_{tot}}{N_B} (f_4 + f_5)$$
 (F.6)

The average discharge burnup is the assembly weighted average of B_{dis,A} and B_{dis,B}; namely:

$$B_{dis,avg} = \frac{N_A B_{dis,A} + N_B B_{dis,B}}{N_A + N_B}$$
(F.7)

Combining Eqs. F.5 through F.7 gives:

$$B_{dis,avg} = B_C \frac{N_{tot}}{N_A + N_B}$$
(F.8)

In terms of Eq. F.4:

^Bdis,avg =
$$\frac{\rho_{o} - \rho_{L}}{A\left[\frac{N_{A}+N_{B}}{N_{A}}\left[\sum_{i=1}^{S}\sum_{j=i}^{j}f_{i}f_{j}\right] + \frac{N_{A}+N_{B}}{N_{B}}\left[\sum_{j=4}^{S}\sum_{j=i}^{j}f_{i}f_{j}\right]}_{(F.9)}$$

The following constituent equations:

$$N_{reload} = N_{A} + N_{B}$$

$$N_{A} = N_{tot} - 2N_{reload}$$

$$N_{B} = 3N_{reload} - N_{tot}$$

$$N_{reload} = N_{tot} \frac{B_{c}}{B_{dis,avg}}$$
(F.10)

can be combined with Eq. F.9 to give

$${}^{B}_{dis,avg} = \frac{\rho_{o} - \rho_{L}}{A\left[\frac{1}{(B_{dis}/B_{c}-2)}\left[\sum_{i=1}^{3} \sum_{j=i}^{3} f_{i}f_{j}\right] + \frac{1}{(3-B_{dis}/B_{c})}\left[\sum_{i=4}^{5} \sum_{j=i}^{5} f_{i}f_{j}\right]}_{(F.11)}$$

Consider the case $N_A = N_B$, which gives $B_{dis}/B_c = 2.5$. Then the average discharge burnup becomes:

$$B_{\text{dis,avg}} = \frac{\rho_{0} - \rho_{L}}{A \cdot 2 \left[\begin{bmatrix} 3 & 3 \\ \sum & \sum & f_{i}f_{j} \end{bmatrix} + \begin{bmatrix} 5 & 5 \\ \sum & \sum & f_{i}f_{j} \end{bmatrix}} \right]$$
(F.12)

and the CSI (cycle schedule index) is:

$$CSI = 2\left\{ \begin{bmatrix} 3 & 3 \\ \sum & \sum \\ i=1 & j=i \end{bmatrix} + \begin{bmatrix} 5 & 5 \\ \sum & \sum \\ i=4 & j=i \end{bmatrix} \right\}$$
(F.13)

The cycle schedule index for equal power sharing $(f_1 = f_2 = f_3 = f_4 = f_5)$ is 0.72. However, the minimum cycle schedule index (which maximizes burnup and minimizes natural uranium usage) occurs when $f_1 = f_2 = f_3$ and $f_4 = f_5$, but the two sets of f_1 's are not equal. For this case $(N_A = N_B)$ the minimum occurs when:

$$f_1 = f_2 = f_3 = 0.1766$$

 $f_4 = f_5 = 0.2350$

and the CSI is 0.7059, 2% lower than the equal-power-sharing case. Recall, that for integer batch numbers (N=2,3,4...) with equal power sharing the CSI was equal to $\frac{N+1}{2N}$. One might be tempted to apply this relation to the non-integer problem of present concern: N=2.5, in which case $\frac{N+1}{2N} = 0.700$, about 0.85% lower than the true optimum (and hence physically unreachable). This example reinforces the observation that an accurate computation of the cycle schedule index is essential to determine discharge burnups and hence uranium savings, particularly when one is trying to distinguish among subtle variations in design or fuel management strategy.

APPENDIX G

MATHEMATICAL EQUIVALENCE OF COOLANT VOID SPECTRAL SHIFT WITH SPECTRAL SHIFT BY PIN-PULLING AND BUNDLE RECONSTITUTION

Spectral shift in a reactor can be accomplished by pitch variation (corresponding to pin-pulling and bundle reconstitution) or by varying the void content of the coolant. While slight differences may occur in the ρ vs. B values for the two systems because of the difference in Dancoff shadowing factors, the two approaches can be made identical for all practical purposes.

Consider spectral shift accomplished by pin-pulling and bundle reconstitution (see Fig. 7-10).

Suppose the reactor contains N_1 assemblies of batch one fuel, each assembly containing M' metric tons of heavy metal and N_2 (N_3) assemblies of batch 2 (3) fuel, each assembly containing M" metric tons of heavy metal. The spectral shift unit cell (or assembly) calculation determines the values of ρ'_0 , A' and A", where ρ'_0 is the initial extrapolated reactivity of the "dry" assembly, A' is the slope of the reactivity vs. burnup curve (MWD/MT)⁻¹ for the "dry" assembly and A" is the slope of the ρ vs. burnup curve (MWD/MT)⁻¹ for the "wet" assembly. The reactivity equations for the assemblies in the three batches can be written as:

$$\rho_{1} = \rho_{0}' - \frac{A'(P_{avg}t) P_{1}}{M'}$$

$$\rho_{2} = \rho_{0}' - \frac{A'(P_{avg}t)P_{1}}{M'} - \frac{A''(P_{avg}t)P_{2}}{M''} + \Delta \rho_{\omega}$$

$$\rho_{3} = \rho_{0}' - \frac{A'(P_{avg}t)P_{1}}{M'} - \frac{A''(P_{avg}t)P_{2}}{M''} - \frac{A''(P_{avg}t)P_{3}}{M''} + \Delta \rho_{\omega}$$
(G.1)

where P_{avg} is the average assembly power, P_i is the relative power in assemblies of type i, and the product P_{avg} t can be identified as the average energy generated per assembly in one cycle. The power fraction for each batch is given by:

$$f_1 = \frac{N_1 P_1}{N_{tot}}$$
 $f_2 = \frac{N_2 P_2}{N_{tot}}$ $f_3 = \frac{N_3 P_3}{N_{tot}}$ (G.2)

Applying the criticality equation:

$$\rho_{\mathbf{L}} = \sum_{i=1}^{3} \mathbf{f}_{i} \rho_{i}$$
 (G.3)

gives

$$\overline{P}t = \frac{\rho_{o}' - \rho_{L} + \Delta \rho_{\omega} \left[\frac{N_{2}P_{2} + N_{3}P_{3}}{N_{tot}}\right]}{\frac{A'P_{1}}{M'} + \frac{A''}{M''} \left[\frac{N_{2}P_{2}^{2}}{N_{tot}} + \frac{N_{3}P_{3}^{2}}{N_{tot}} + \frac{N_{3}P_{3}^{2}}{N_{tot}}\right]}$$
(G.4)

or:

$$\overline{P}t = \frac{\rho_{0}' - \rho_{L} + \Delta \rho_{\omega} (f_{2}+f_{3})}{\frac{A'}{M'} \frac{N_{tot}}{N_{1}} f_{1} + \frac{A''}{M''} \left[\frac{N_{tot}}{N_{2}} f_{2}^{2} + \frac{N_{tot}}{N_{2}} f_{2}f_{3} + \frac{N_{tot}}{N_{3}} f_{3}^{2} \right]$$
(G.5)

The cycle burnups for each batch are given by:

$$B_{c1} = \frac{\overline{P}t P_1}{M'}; B_{c2} = \frac{\overline{P}t P_2}{M''}; B_{c3} = \frac{\overline{P}t P_3}{M''}$$
 (G.6)

The cycle average burnup is the mass-weighted value of the batch burnups, hence:

$$B_{c,avg} = \frac{N_1 M'(\frac{\overline{Pt}}{M'})P_1 N_2 M''(\frac{\overline{Pt}}{M''})P_2 N_3 M''(\frac{\overline{Pt}}{M''})P_3}{M_{total}}$$
(G.7)

and so:

$$B_{c,avg} = \frac{\overline{Pt} N_{tot}}{M_{tot}}$$
(G.8)

In the ideal case, pins are pulled and the batch two bundles are manufactured immediately during the refueling outage, to permit reinsertion of the reconstituted bundles before startup. (Actually this is not essential: one could postpone reinsertion one cycle and obtain the same <u>steady-state</u> condition.) All mass is preserved, so that $N_1M' = N_2M'' = N_3M''$. The reload mass is N_1M'' and the reload feed (MTF) is $(\frac{F}{P}) N_1M'$ where F/P is the feed to product ratio. The total energy generated per cycle is given by:

$$E = B_{c,avg} (M_{tot}) = \overline{P}t N_{tot}$$
(G.9)

and the total energy per metric ton of feed is given by:

$$U = \frac{E}{F} = \frac{\overline{Pt} N_{tot}}{(\frac{F}{P}) N_{\underline{1}}M'}$$
(G.10)

where U is the uranium utilization (MWD/MTF). Inserting Eq. G.5 and making use of the fact that $N_1M' = N_2M'' = N_3M''$ gives:

$$U = \frac{\rho_{o}' - \rho_{L} + \Delta \rho_{\omega} (f_{2} + f_{3})}{[A'f_{1} + A''(f_{2}^{2} + f_{2}f_{3} + f_{3}^{2})] (\frac{F}{P})}$$
(G.11)

Eq. G.ll is identical to the corresponding relation developed for coolant spectral shift in Chapter 7. As shown in Chapter 7, the choice of transition burnup, B_{tran} , specifies f_1 (P_1), and hence the optimum uranium utilization occurs when $f_2 = f_3$, which in the pin-pulling and bundle reconstitution case corresponds to $P_2 = P_3$. Optimizing for f_1, f_2, f_3 in the coolant spectral shift case (and, hence, P_1, P_2, P_3 in the pin-pulling case) does not imply that such optimal power profiles can be achieved in actual practice.

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