FF"

Huar

Introduction

Naive

Point-Value Complex Roots Iterative

Application

2D Convolutions Separable Kernel

Conclusion

Sample

Past Lecture

Poforoncos

Fast Fourier Transform and 2D Convolutions Doing convolutions fast

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TJ Vision & Graphics Club, October 21, 2020

Table of Contents

FFT

Huan

Introduction

Algorithms

Naive

Point-Value Complex Root

NTT

Application

Audio 2D Convolutions Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

Introduction

- 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - Separable Kerrieis
 - FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- 6 Past Lectures
- 7 References

Introduction

FF1

Huai

Introduction

Algorithms

Naive

Point-Value Complex Root Iterative

Applications

Audio

2D Convolution Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

The Fast Fourier Transform (FFT) is a common technique for signal processing and has many engineering applications. It also has a fairly deep mathematical basis, but we will ignore both those angles in favor of accessibility. Instead, we will approach the FFT from the most intuitive angle, polynomial multiplication.

Introduction

FF1

Huai

Introduction

Algorithm

Point-Value Complex Roo Iterative

Applications
Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

The Fast Fourier Transform (FFT) is a common technique for signal processing and has many engineering applications. It also has a fairly deep mathematical basis, but we will ignore both those angles in favor of accessibility. Instead, we will approach the FFT from the most intuitive angle, polynomial multiplication.

First, we represent polynomials by a list of coefficients, where the number at index 0 represents the coefficient of x^0 , the number at index 1 represents the coefficient of x^1 , and so on.

Polynomial representation

The polynomial $3 + 2x + 4x^2$ becomes [3, 2, 4].

Definition of the Convolution

FFT

Huai

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Applications

2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

The multiplication of two polynomials f and g is then simply each term of f multiplied with each term of g and then added up. We can also assume that f and g are the same length N, where the polynomial of lesser degree is padded with zeros. If we say the product is p, we can give an formula for an index in p in the following way:

Definition of the convolution

$$p[n] = (f * g)[n] = \sum_{i=0}^{n} f[i]g[n-i]$$

Intuition Behind a Convolution

FF1

Huar

Introduction

Algorithm

FTT
Point-Value
Complex Roo

Complex Roo Iterative NTT

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

p[n] is the coefficient of x^n in the product, and it is formed by adding up all the possible ways to get to x^n , i.e. $f[0]x^0$ times $g[n]x^n$, $f[1]x^1$ times $g[n-1]x^{n-1}$, etc. Intuitively, this "flips" g, and then the resulting product is computed by "sliding" g over f and then computing the dot product between the two lists, or a weighted average.

FF1

Huar

Introduction

Algorithms

Naive

Point-Value Complex Root Iterative

Application

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

FF1

Huar

Introduction

Algorithms

Point-Value
Complex Root

Applications

2D Convolutions
Separable Kerne

onclusion

Sample Problem

Past Lecture

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1]. We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

FF1

Huar

Introduction

Algorithms

Point-Value Complex Root Iterative

NTT Applications

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

FF"

Huar

Introduction

Algorithms

Point-Value Complex Roo Iterative

Applications

2D Convolutions
Separable Kerne
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

Then, [3, 1] overlaps with [3, 2] so we get $3 \cdot 3 + 1 \cdot 2 = 11$.

FF"

Huar

Introduction

Algorithm:

Point-Value
Complex Roo
Iterative
NTT

Applications Audio

2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

Then, [3, 1] overlaps with [3, 2] so we get $3 \cdot 3 + 1 \cdot 2 = 11$.

 $[2,\,3,\,1]$ overlaps with $[3,\,2,\,4]=16$

FF"

Huar

Introduction

Algorithms

Point-Value
Complex Roo
Iterative
NTT

Applications Audio

2D Convolutions Separable Kernels FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

Then, [3, 1] overlaps with [3, 2] so we get $3 \cdot 3 + 1 \cdot 2 = 11$.

[2, 3, 1] overlaps with [3, 2, 4] = 16

[2, 3] overlaps [2, 4] = 16,

FF"

Huar

Introduction

Algorithm:

Point-Value
Complex Root
Iterative
NTT

Applications Audio

2D Convolutions Separable Kernels FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

Then, [3, 1] overlaps with [3, 2] so we get $3 \cdot 3 + 1 \cdot 2 = 11$.

[2, 3, 1] overlaps with [3, 2, 4] = 16

[2, 3] overlaps [2, 4] = 16,

and finally [2] overlaps with [4] to give 8.

FF"

Huar

Introduction

Algorithm:

Point-Value
Complex Roo
Iterative
NTT

Applications Audio

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial $3 + 2x + 4x^2 = [3, 2, 4]$ and the polynomial $1 + 3x + 2x^2 = [1, 3, 2]$. To compute their product, we first flip the second list to get [2, 3, 1].

We then slide [2, 3, 1] over [3, 2, 4], imagining there are zeros such that the parts of [2, 3, 1] that don't overlap with [3, 2, 4] aren't counted.

For the first value, 1 overlaps with 3 so we get 3.

Then, [3, 1] overlaps with [3, 2] so we get $3 \cdot 3 + 1 \cdot 2 = 11$.

[2, 3, 1] overlaps with [3, 2, 4] = 16

[2, 3] overlaps [2, 4] = 16,

and finally [2] overlaps with [4] to give 8.

Our final answer is then [3, 11, 16, 16, 8] =

$$3 + 11x + 16x^2 + 16x^3 + 8x^4 = (3 + 2x + 4x^2)(1 + 3x + 2x^2).$$

Commutativity of the Convolution

FF1

Huar

Introduction

Algorithms

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

Deferences

What if we computed g * f? It should be the same since polynomial multiplication should be commutative.

Theorem

f*g=g*f, i.e. polynomial multiplication is commutative.

Proof of the Commutativity of the Convolution

FFT

Huar

Introduction

Algorithms

Point-Value
Complex Roo

Applications

2D Convolutions Separable Kerne

onclusion

Sample Problem

Past Lectures

References

Proof.

We have $(f * g)(n) = \sum_{i=0}^n f[i]g[n-i]$ by definition. Perform the variable substitution k = n - i, so i = n - k. Summing from $\sum_{i=0}^n$ will sum from k = n to k = 0 in descending order, so $\sum_{i=0}^n = \sum_{k=0}^n$ (from the commutativity of addition).

$$(f * g)[n] = \sum_{i=0}^{n} f[i]g[n-i]$$
 Definition
$$= \sum_{k=0}^{n} f[n-k]g[k]$$
 Substitution
$$= (g * f)$$

The Fast Fourier Transform

FF1

Huar

Introduction

Algorithm

Point-Value
Complex Root
Iterative

Applications Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lecture:

References

This operation is known as a convolution, which is equivalent to polynomial multiplication in the discrete case and is denoted f * g. Its relevance to image processing will be expounded on later (for now, this puts the "convolutional" in "Convolutional Neural Networks"). Today's lecture is about the Fast Fourier Transform, an efficient algorithm to perform convolutions.

Table of Contents

FFT

Huan

Introduction

A.1.

Naive

FTT

Complex Root

итт

Application

2D Convolutions Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

- Introduction
- 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - Separable Kerrieis
 - FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- 6 Past Lectures
- 7 References

Naive

FF1

Huai

Introduction

Algorithms

Naive FTT

Complex Roots

Application

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

A naive approach to the convolution of two lists of length N, M will have runtime O(NM) using the standard polynomial multiplication algorithm (each term of the first list multiplied with each term of the second list).

Naive

FF1

Huai

Introduction

Algorithm:
Naive
FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

A naive approach to the convolution of two lists of length N, M will have runtime O(NM) using the standard polynomial multiplication algorithm (each term of the first list multiplied with each term of the second list).

What will be the length of the resulting list?

FF1

Huai

Introduction

Algorithm
Naive

Point-Value Complex Roots Iterative

Applications Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

A naive approach to the convolution of two lists of length N, M will have runtime O(NM) using the standard polynomial multiplication algorithm (each term of the first list multiplied with each term of the second list).

What will be the length of the resulting list? The first list is a polynomial of degree N-1, the second of degree M-1, so the resulting polynomial has degree (N-1)+(M-1)=N+M-2.

Naive

Naive

A naive approach to the convolution of two lists of length N, M will have runtime O(NM) using the standard polynomial multiplication algorithm (each term of the first list multiplied with each term of the second list).

What will be the length of the resulting list?

The first list is a polynomial of degree N-1, the second of degree M-1, so the resulting polynomial has degree

$$(N-1)+(M-1)=N+M-2.$$

A polynomial of degree D has D+1 coefficients, so the length of the product is N + M - 1. Thus,

FF"

Huai

Introduction

Algorithm:
Naive
FTT
Point-Value

Application

2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

A naive approach to the convolution of two lists of length N, M will have runtime O(NM) using the standard polynomial multiplication algorithm (each term of the first list multiplied with each term of the second list).

What will be the length of the resulting list? The first list is a polynomial of degree N-1, the second of degree M-1, so the resulting polynomial has degree

(N-1)+(M-1)=N+M-2.

A polynomial of degree D has D+1 coefficients, so the length of the product is N+M-1. Thus,

Lower bound for the runtime of a convolution

The length of the convolution is N + M - 1, which is linear, so a better runtime than quadratic could exist.

Table of Contents

FFT

Huan

Introduction

Algorithma

Naive

Point-Value Complex Root Iterative

NTT

Applications

Audio

2D Convolutions
Separable Kernel:
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

- Introduction
- 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- 6 Past Lectures
- 7 References

Point-Value Representation

An alternative way to express a polynomial

FFT

Huar

Introductio

Algorithms

Naive FTT

Point-Value Complex Root Iterative

Applications

Audio
2D Convolutions
Separable Kerne
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

The key observation is that we can represent polynomials in a different form than a coefficient list. In particular, we can use a point-value representation, or a list of (x, y) pairs that give an input and the corresponding output of a polynomial.

Point-value representation

A point-value representation for a polynomial p is a list of distinct x values and their corresponding y values, e.g. $\{(x_0, p(x_0)), (x_1, p(x_1)), \dots, (x_n, p(x_n))\}$

Moving between Representations

FF1

Huai

Introduction

Algorithms

Naive

Point-Value Complex Roots Iterative

Applications

2D Convolution Separable Kern

onclusion

Sample Problem

Past Lectures

Deferences

Evaluation

We call the process of going from a coefficient representation to a point-value representation *evaluation*, since we evaluate the polynomial at multiple points to get the point-value representation.

Moving between Representations

FF1

Huar

Introductio

Algorithms

Naive FTT

Point-Value
Complex Root
Iterative

Applications

2D Convolutions
Separable Kerne
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

Deferences

Evaluation

We call the process of going from a coefficient representation to a point-value representation *evaluation*, since we evaluate the polynomial at multiple points to get the point-value representation.

Interpolation

Likewise, we call the process of going from a point-value representation to a coefficient representation *interpolation*, since we are finding a polynomial which "fits" the data.

Moving between Representations

FFT

Huar

Introductio

Algorithm

Naive FTT Point-Value

> Complex Root Iterative NTT

Applications

Audio

Audio
2D Convolutions
Separable Kerne
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Evaluation

We call the process of going from a coefficient representation to a point-value representation *evaluation*, since we evaluate the polynomial at multiple points to get the point-value representation.

Interpolation

Likewise, we call the process of going from a point-value representation to a coefficient representation *interpolation*, since we are finding a polynomial which "fits" the data.

Suppose we have a polynomial of degree n. We then need a certain number of points for evaluation and interpolation to be well-defined.

Well-defined Representations

FF1

Huai

Introduction

Algorithr

Point-Value Complex Root

Iterative

Application:

2D Convolution Separable Kerne FFT Algorithm

onclusion

Sample Problem

Past Lectures

References

Evaluation is always well-defined, because we can always evaluate a polynomial of any degree or coefficient representation. However, if we don't have enough points, interpolation is not necessarily possible.

Well-defined Representations

FF1

Ниа

Introduction

Algorithm Naive

> Point-Value Complex Root Iterative

Application:

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Evaluation is always well-defined, because we can always evaluate a polynomial of any degree or coefficient representation. However, if we don't have enough points, interpolation is not necessarily possible.

Ill-defined representation

Consider the point-value representation [(0,0),(1,1)] and a degree of 2. This could be the polynomial x^2 or $2x^2 - x$.

Well-defined Representations

EF1

Hua

Introductio

Algorithm

Point-Value Complex Root Iterative

Audio
2D Convolutions
Separable Kerne

Conclusion

Sample Problems

Past Lectures

References

Evaluation is always well-defined, because we can always evaluate a polynomial of any degree or coefficient representation. However, if we don't have enough points, interpolation is not necessarily possible.

Ill-defined representation

Consider the point-value representation [(0,0),(1,1)] and a degree of 2. This could be the polynomial x^2 or $2x^2 - x$.

So for a polynomial of degree n, we need at least n+1 distinct points (since each point gives another linear equation constraining the n+1 coefficients of the polynomial). We can in fact prove that if we have n+1 points, that uniquely determines a polynomial of degree n.

Proof of the Point-Value Representation

FFT

Huai

Introductio

Algorithms

Point-Value Complex Roo

Applications

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

$\mathsf{Theorem}$

A point-value representation with n distinct points uniquely determines a polynomial of degree n-1.

Proof.

We have a polynomial of the form

 $p(x) = a_0 + a_1x + a_2x^2 + \cdots + a_{n-1}x^{n-1}$ and n points of the form (x_i, y_i) such that $p(x_i) = y_i$. Those constraints determine the following matrix equation:

$$\begin{bmatrix} 1 & x_0 & x_0^2 & \cdots & x_0^{n-1} \\ 1 & x_1 & x_1^2 & \cdots & x_1^{n-1} \\ \vdots & \vdots & \vdots & \ddots & \vdots \\ 1 & x_{n-1} & x_{n-1}^2 & \cdots & x_{n-1}^{n-1} \end{bmatrix} \begin{bmatrix} a_0 \\ a_1 \\ \vdots \\ a_{n-1} \end{bmatrix} = \begin{bmatrix} y_0 \\ y_1 \\ \vdots \\ y_{n-1} \end{bmatrix}$$

Proof of the Point-Value Representation

FF1

Huar

Introductio

Algorithms

Naive FTT Point-Value

Complex Roots
Iterative

Applications
Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Proof.

The leftmost matrix is known as the *Vandermonde matrix*, denoted $V(x_0, x_1, \ldots, x_{n-1})$ which has the determinant (left as an exercise for the reader)

$$\prod_{0 \le j < i \le n-1} (x_i - x_j)$$

A matrix is invertible if and only if its determinant is nonzero, so this matrix is invertible if each x_i is distinct. Thus, we can solve for the coefficients by multiplying by the inverse, so $\vec{a} = V^{-1}\vec{y}$, and this solution is unique since an invertible matrix is a bijective transformation between a vector space and itself.

Lagrange's Formula

FFT

Huar

Introductio

......

Naive

FTT Point-Value

Complex Root

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

This proof directly gave an easy construction of the interpolating polynomial, by $V^{-1}\vec{y}$. Matrix inverses can be computed in $O(n^3)$ as an easy upper bound, but that can be improved with Lagrange's interpolating formula to yield a $O(n^2)$ time algorithm. I will not elaborate on Lagrange's formula in this lecture but a good Wikipedia page is available here.

Evaluation

FF1

Huar

Introductio

۸ ا ناجار ۱

Naive

Point-Value

Complex Roots
Iterative

Application

2D Convolutions
Separable Kernel

 $\mathsf{Conclusion}$

Sample Problem

Past Lecture

References

If we have a list of N coefficients, then the polynomial is of degree N-1 and thus we need N distinct points. We first figure out how to evaluate polynomial at a single point, and will repeat the process for all the points.

Evaluating a Polynomial

FF1

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Applications

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

Deferences

Suppose we have a polynomial of the form $p(x) = a_0 + a_1x + a_2x^2 + \cdots + a_{n-1}x^{n-1}$. If we evaluate at a particular x_0 , we compute each $a_ix_0^i$ term which would take $O(N^2)$ time with repeated multiplication and $O(N \log N)$ time with fast exponentiation.

Horner's Rule

FF1

Huar

Introductio

Algorithm Naive

Naive
FTT
Point-Value

Complex Roots

Applications

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

But we can do better with Horner's rule. We notice that the degree in coefficient form is monotonically increasing, so we can successively factor out a multiplication by x.

Horner's rule

$$p(x) = a_0 + x(a_1 + x(a_2 + \cdots + x(a_{n-2} + xa_{n-1})))$$

Horner's Rule

FF1

Huar

Introductio

Algorithm

Point-Value Complex Roots Iterative

Applications
Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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Horner's rule

$$p(x) = a_0 + x(a_1 + x(a_2 + \cdots + x(a_{n-2} + xa_{n-1})))$$

We do N multiplications and N additions, so the algorithm runs in O(N). Evaluating a polynomial at N points then takes $O(N \cdot N) = O(N^2)$ time.

FF1

Huar

Introductio

Algorithm

FTT

Point-Value

Complex Roots
Iterative

Application

2D Convolutions Separable Kernel

onclusion

Sample Problem

Past Lecture

References

So we can do both evaluation and interpolation in $O(n^2)$ and both are well-defined if we have enough points. Why did we figure this out? We can multiply two polynomials efficiently if we have the point-value representations of each!

Multiplying in Linear Time

FFT

Huai

Introduction

Algorithms

Naive

Point-Value Complex Roots

Iterative NTT

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Suppose we have polynomials f, g in coefficient form. We also assume that the polynomials are evaluated at the same points, so we have

$$[(x_0, f(x_0)), (x_1, f(x_1)), \dots]$$

and

$$[(x_0,g(x_0)),(x_1,g(x_1)),\ldots]$$

f * g is then simply

$$[(x_0, f(x_0)g(x_0)), (x_1, f(x_1)g(x_1)), \ldots]$$

or the element-wise multiplication of the two lists.

This can be easily computed in O(n)!

Summary

FF1

Huar

Introductio

Naive

Point-Value

Complex Roots Iterative

Application

Audio
2D Convolutions
Separable Kernel

Conclusion

Sample Problem

Past Lecture

Deferences

So our algorithm for polynomial multiplication is as follows:

Evaluate a coefficient representation into a point-value representation.

Summary

FF1

Huar

Introductio

Naive FTT

Point-Value

Complex Roots Iterative

Application

2D Convolutions
Separable Kernels

onclusion

Sample Problem

Past Lecture

Deferences

So our algorithm for polynomial multiplication is as follows:

- Evaluate a coefficient representation into a point-value representation.
- 2 Multiply the two point-value representations in linear time.

Summary

FF"

Huar

Introductio

....

Naive FTT

FTT Point-Value

Complex Roots
Iterative

Applications

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

Deferences

So our algorithm for polynomial multiplication is as follows:

- Evaluate a coefficient representation into a point-value representation.
- 2 Multiply the two point-value representations in linear time.
- Interpolate the resulting point-value representation back to coefficients.

FF

Huar

Introduction

Algorithn

FTT

Point-Value Complex Root

NTT

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

The speed of this algorithm is contingent on our ability to quickly evaluate and interpolate a polynomial. Currently, with our $O(n^2)$ time evaluation and interpolation algorithms we match the $O(n^2)$ naive algorithm. However, under this framework we can improve the time if we pick our points cleverly rather than arbitrarily.

Complex Roots of Unity

FF1

Huar

Introductio

Algorithm

Naive

Point-Value
Complex Roots
Iterative

Application

2D Convolutions Separable Kernels FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

Our special points are going to be complex roots of unity, or roots of 1 that are allowed to have an imaginary component.

Complex roots of unity

The second root of 1 can be 1 or -1 (taking "second root" to mean anything which squared is 1). The fourth root of 1 can be 1, -1, i, or -i. (since $i^4 = (i^2)^2 = (-1)^2 = 1$).

Computing Complex Roots

Complex Roots

To easily compute these roots, we can rewrite 1 using

Euler's formula

 $e^{ix} = \cos x + i \sin x$ (a proof of this appears in the appendix).

Computing Complex Roots

FFT

Huai

Introduction

Algorithm Naive FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolution Separable Kerne FFT Algorithm

Conclusior

Sample Problem

Past Lectures

Deferences

To easily compute these roots, we can rewrite 1 using

Euler's formula

 $e^{ix} = \cos x + i \sin x$ (a proof of this appears in the appendix).

$$e^{2\pi i} = \cos 2\pi + i \sin 2\pi = 1$$

Computing Complex Roots

FFT

Huai

Introduction

Algorithm
Naive
FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

To easily compute these roots, we can rewrite 1 using

Euler's formula

 $e^{ix} = \cos x + i \sin x$ (a proof of this appears in the appendix).

$$e^{2\pi i} = \cos 2\pi + i \sin 2\pi = 1$$

So we can take a *n*th root by simply raising

$$\sqrt[n]{1} = (e^{2\pi i})^{\frac{1}{n}}$$

so a root is

$$e^{\frac{2\pi i}{n}}$$

FFT

Huar

Introduction

A | --- -- :-- |

Algorithm.

FTT

Point-Value Complex Roots

Iterative

NTT

Application

2D Convolutions
Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

However, note that we can rewrite 1 in many different ways since sine and cosine are periodic.

FF1

Huar

Introduction

Algorithm Naive

Point-Value
Complex Roots
Iterative

Application Audio

2D Convolution Separable Kerne

Conclusion

Sample Problem

Past Lectures

Deferences

However, note that we can rewrite 1 in many different ways since sine and cosine are periodic.

Since adding 2π doesn't change the value of sine and cosine, 1 is also equal to $e^{4\pi i}$, $e^{6\pi i}$, and so on.

FF"

Huai

Introduction

Algorithm Naive FTT

Point-Value
Complex Roots
Iterative
NTT

Applications

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

Deferences

However, note that we can rewrite 1 in many different ways since sine and cosine are periodic.

Since adding 2π doesn't change the value of sine and cosine, 1 is also equal to $e^{4\pi i}$, $e^{6\pi i}$, and so on.

In general, $e^{2\pi ki}$ is equal to 1 for any integer k, so if we take the nth root, $e^{\frac{2\pi ki}{n}}$ is also going to be a valid root of unity.

FF1

Huai

Introductio

Algorithm
Naive
FTT
Point-Valu

Complex Roots
Iterative
NTT

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

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Since adding 2π doesn't change the value of sine and cosine, 1 is also equal to $e^{4\pi i}$, $e^{6\pi i}$, and so on.

In general, $e^{2\pi ki}$ is equal to 1 for any integer k, so if we take the nth root, $e^{\frac{2\pi ki}{n}}$ is also going to be a valid root of unity. However, not every k gives a distinct root of unity.

k = n + 1 is equivalent to k = 1 since

$$\cos\left(\frac{2\pi(n+1)}{n}\right) = \cos\left(2\pi + \frac{2\pi}{n}\right) = \cos\left(\frac{2\pi}{n}\right)$$

This generalizes such that an power k equivalent to $j \mod n$ will have the same root.

Principle Root of Unity

FF1

Huar

Introduction

Naive FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We can easily keep track of the n distinct nth roots of unity by writing them as powers of the principle root of unity.

Principle Root of Unity

The *principle root of unity* is the root of unity when k = 1.

Principle Root of Unity

FFT

Huai

Introduction

.....

Naive FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We can easily keep track of the n distinct nth roots of unity by writing them as powers of the principle root of unity.

Principle Root of Unity

The *principle root of unity* is the root of unity when k = 1.

Notation

We will denote this principle root as ω_n , where $\omega_n = e^{\frac{2\pi i}{n}}$.

Principle Root of Unity

FF1

Huai

Introduction

Algorithm
Naive

Point-Value
Complex Roots
Iterative

Applications
Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

We can easily keep track of the *n* distinct *n*th roots of unity by writing them as powers of the principle root of unity.

Principle Root of Unity

The *principle root of unity* is the root of unity when k = 1.

Notation

We will denote this principle root as ω_n , where $\omega_n=e^{\frac{2\pi i}{n}}$.

Since we picked k = 1, we can represent every nth root of unity as a power of this root of unity since

$$e^{\frac{2\pi ki}{n}} = \left(e^{\frac{2\pi i}{n}}\right)^k = \omega_n^k$$

Properties of a Principle Root

FFT

Huan

Introductio

Algorithms

Naive

Point-Value Complex Roots

Iterative NTT

Application

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Note that every power of the principle root of unity is itself a root of unity, because

$$(\omega_n^k)^n = (\omega_n^n)^k = 1^k = 1$$

Properties of a Principle Root

FFT

Huar

Introductio

Algorithm

Point-Value
Complex Roots

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Note that every power of the principle root of unity is itself a root of unity, because

$$(\omega_n^k)^n = (\omega_n^n)^k = 1^k = 1$$

We now come to an observation that will be instrumental in developing the FFT - that the square of a nth principle root of unity is a $\frac{n}{2}$ th principle root of unity.

This follows nearly from definition:

$$\omega_n^2 = \left(e^{\frac{2\pi i}{n}}\right)^2$$

$$= e^{\frac{4\pi i}{n}}$$

$$= e^{\frac{2\pi i}{2}}$$

$$= \omega_n$$

FF1

Huar

Introductio

Algorithm

FTT

Complex Roots

Iterative

Application

2D Convolutions
Separable Kernel

onclusion

Sample Problem

Past Lecture

References

We now show that evaluating a polynomial at *n* distinct *n*th roots of unity can be written as a recurrence relation. Our observation is a clever rewrite of a polynomial into two parts.

Splitting a Polynomial

FFT

Huai

Introduction

Naive FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolution Separable Kern FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial

$$p(x) = a_0 + a_1x + a_2x^2 + \cdots + a_{n-1}x^{n-1}$$

We divide the coefficient list of p into two parts, one with even powers and the other with odd powers, the left and right halves respectively.

Splitting a Polynomial

FF1

Huai

Introduction

Algorithm Naive FTT

Point-Value
Complex Roots
Iterative

Applications
Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Suppose we have the polynomial

$$p(x) = a_0 + a_1x + a_2x^2 + \cdots + a_{n-1}x^{n-1}$$

We divide the coefficient list of p into two parts, one with even powers and the other with odd powers, the left and right halves respectively.

We assume that n is a power of 2 so that p can always be divided in such a manner.

If n isn't, we can always pad with 0's.

Splitting a Polynomial

FF1

Huai

Introductio

Algorithi Naive FTT

Point-Value
Complex Roots
Iterative

Application

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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We divide the coefficient list of p into two parts, one with even powers and the other with odd powers, the left and right halves respectively.

We assume that n is a power of 2 so that p can always be divided in such a manner.

If n isn't, we can always pad with 0's.

$$p(x) = a_0 + a_1 x + a_2 x^2 + \dots + a_{n-1} x^{n-1}$$
 (1)

$$L(x) = a_0 + a_2 x + a_4 x^2 + \dots$$
(2)

$$R(x) = a_1 + a_3 x + a_5 x^2 + \dots$$
(3)

It follows that p can be written in terms of L and R:

$$p(x) = L(x^2) + xR(x^2)$$
 (4)

EF1

Huan

Introduction

Naive

FTT

Complex Roots
Iterative

Application .

Audio 2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Recall that we are trying to evaluate p at n roots of unity.

FFT

Huar

Introduction

Algorithm

Naive

Point-Value Complex Roots Iterative

Application

2D Convolution: Separable Kerne

onclusion

Sample Problem

Past Lecture

References

Recall that we are trying to evaluate p at n roots of unity. Suppose we have a function that takes as input a list of coefficients and returns the evaluation at n roots of unity.

FF1

Huar

Introduction

Algorithm

Naive

Point-Value

Complex Roots

Iterative

NTT

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Recall that we are trying to evaluate p at n roots of unity. Suppose we have a function that takes as input a list of coefficients and returns the evaluation at n roots of unity. We can define this function in terms of itself, because we have a recurrence relation - divide the list in two, giving us L evaluated at $\frac{n}{2}$ th roots of unity and the same for R (from the fact that a nth root of unity squared is a $\frac{n}{2}$ th root of unity).

FF1

Huar

Introductio

Algorithm

Point-Value
Complex Roots
Iterative

Audio
2D Convolutions
Separable Kernels

Conclusion

Sample Problem

Past Lectures

References

Recall that we are trying to evaluate p at n roots of unity. Suppose we have a function that takes as input a list of coefficients and returns the evaluation at n roots of unity. We can define this function in terms of itself, because we have a recurrence relation - divide the list in two, giving us L evaluated at $\frac{n}{2}$ th roots of unity and the same for R (from the fact that a nth root of unity squared is a $\frac{n}{2}$ th root of unity). Finally, we can reconstruct p from L and R according to (4).

Accounting for Edge Cases

FFT

Huar

Introductio

Naive

Point-Val

Complex Roots
Iterative

NTT

Applications

2D Convolutions
Separable Kerne

onclusion

Sample Problem

Past Lecture

Deferences

This works directly for ω_n^0 to $\omega_n^{\frac{n}{2}-1}$, however for a power greater than $\frac{n}{2}-1$ we need to put it in terms of a power less than $\frac{n}{2}$ (since L and R are only $\frac{n}{2}$ long).

Derivation of Negative Property

FET

Huan

Introduction

Naive FTT

Point-Valu

Complex Roots
Iterative

Application

Audio

Separable Kerne

Conclusion

Sample Problem

Past Lecture

Deferences

Luckily,

$$\omega_n^{k+\frac{n}{2}} = \cos\left(2\pi \frac{k+\frac{n}{2}}{n}\right) + i\sin\left(2\pi \frac{k+\frac{n}{2}}{n}\right)$$
$$= \cos\left(\frac{2\pi k}{n} + \pi\right) + i\sin\left(\frac{2\pi k}{n} + \pi\right)$$
$$= -\omega_n^k$$

Putting it All Together

FFT

Huar

Introduction

Naive

Point-Valu

Complex Roots

Iterative NTT

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

So, for some power k of the base root of unity we can compute

$$p(\omega_n^k) = L(\omega_n^{2k}) + \omega_n^k R(\omega_n^{2k})$$

and, using the negative property derived before,

$$p(\omega_n^{k+\frac{n}{2}}) = L(\omega_n^{2k}) - \omega_n^k R(\omega_n^{2k})$$

FFT

Huai

Introduction

Naive

FTT

Complex Roots

Iterative NTT

Application:

2D Convolutions Separable Kernel FFT Algorithm

onclusion

Sample Problems

Past Lectures

References

We compute *L* and *R* recursively, and we're done! However, we need to have a base case for the function to work.

FF1

Hua

Introduction

Algorithm

FTT
Point-Value

Complex Roots
Iterative

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lectures

References

We compute L and R recursively, and we're done! However, we need to have a base case for the function to work. Recall our function takes in a coefficient list of length n representing a polynomial, and returns the evaluation of that polynomial at each of the n distinct nth roots of unity.

FF1

Hua

Introduction

Algorithm

FTT
Point-Value
Complex Roots
Iterative

Application

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We compute L and R recursively, and we're done! However, we need to have a base case for the function to work. Recall our function takes in a coefficient list of length n representing a polynomial, and returns the evaluation of that polynomial at each of the n distinct nth roots of unity. The simplest base case is probably n=1, at which we can stop dividing the list in half and just evaluate the polynomial directly.

FF1

Hua

Introduction

Algorithn

Point-Value
Complex Roots
Iterative

Application: Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

We compute L and R recursively, and we're done! However, we need to have a base case for the function to work. Recall our function takes in a coefficient list of length n representing a polynomial, and returns the evaluation of that polynomial at each of the n distinct nth roots of unity. The simplest base case is probably n=1, at which we can stop dividing the list in half and just evaluate the polynomial directly. The only 1st root of unity is 1, and evaluating a polynomial at x=1 is equal the sum of the coefficients, which for a polynomial with one coefficient is just its singular coefficient.

Base Case

FF1

Hua

Introductio

Algorithn
Naive
FTT
Point-Value

Complex Roots
Iterative
NTT

Application Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

We compute L and R recursively, and we're done! However, we need to have a base case for the function to work. Recall our function takes in a coefficient list of length n representing a polynomial, and returns the evaluation of that polynomial at each of the *n* distinct *n*th roots of unity. The simplest base case is probably n = 1, at which we can stop dividing the list in half and just evaluate the polynomial directly. The only 1st root of unity is 1, and evaluating a polynomial at x=1 is equal the sum of the coefficients, which for a polynomial with one coefficient is just its singular coefficient.

Base case

Thus, we can just return the coefficient list of the polynomial, or the input to the function when n = 1.

Analysis of Runtime

FFT

Huar

Introduction

Algorithm

Naive

Point-Value Complex Roots

Iterative

Application

2D Convolution Separable Kerne

onclusion

Sample Problems

Past Lectures

References

At every level of the recursion we divide the list in half, making the depth of the recursive tree $\log n$.

Analysis of Runtime

FFT

Huar

Introductio

Algorithm

Naive

Point-Value
Complex Roots

Iterative

NTT

Applications Audio

2D Convolutions
Separable Kerne

Conclusior

Sample Problem

Past Lectures

References

At every level of the recursion we divide the list in half, making the depth of the recursive tree $\log n$.

At a particular level k, we have 2^k nodes and each node has a list of length $\frac{n}{2^k}$, so the total cost of merging the lists together on that level is $2^k \frac{n}{2^k} = n$.

Analysis of Runtime

FFT

Huar

Introductio

Algorithm

Naive

Point-Value
Complex Roots

Iterative

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

At every level of the recursion we divide the list in half, making the depth of the recursive tree $\log n$.

At a particular level k, we have 2^k nodes and each node has a list of length $\frac{n}{2^k}$, so the total cost of merging the lists together on that level is $2^k \frac{n}{2^k} = n$.

Thus, $O(\log n \cdot n) = O(n \log n)$.

This algorithm has the same recursive pattern as merge sort.

FFT

Huar

Introduction

Algorithms

Naive FTT

Complex Roots

Iterative

Application

Audio 2D Convolutions Separable Kernel: FFT Algorithm

onclusion

Sample Problems

Past Lectures

References

So we can evaluate a polynomial at n roots of unity in $O(n \log n)$ with the above algorithm, called the FFT.

FF1

Huar

Introductio

Algorithms

Naive

Point-Value Complex Roots

\nnlication

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

So we can evaluate a polynomial at n roots of unity in $O(n \log n)$ with the above algorithm, called the FFT. If we want to multiply two polynomials f and g, we can compute $\mathsf{FFT}(f) \circ \mathsf{FFT}(g)$, where \circ is the element-wise multiplication of the outputs in the point-value representations.

FF1

Huar

Introduction

Algorithms

Naive

Point-Value
Complex Roots
Iterative

Application

Audio

2D Convolutions

Separable Kerne

FFT Algorithm

onclusion

Sample Problem

Past Lectures

References

How do we interpolate coefficients from this point-value representation to complete our convolution?

FF1

Huar

Introductio

Algorithm

Naive

Point-Value
Complex Roots
Iterative

Applications Audio

2D Convolutions
Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

How do we interpolate coefficients from this point-value representation to complete our convolution? We need the inverse FFT, which luckily can be written in terms of the FFT. Recall that the FFT essentially computes the multiplication of the Vandermonde matrix with the coefficients to get to the outputs, e.g. $V\vec{a} = \vec{y}$.

FF1

Huar

Introductio

Algorithm

FTT
Point-Value
Complex Roots

Complex Roo Iterative NTT

Audio
2D Convolutions
Separable Kernels

Conclusion

Sample Problem

Past Lecture

References

How do we interpolate coefficients from this point-value representation to complete our convolution? We need the inverse FFT, which luckily can be written in terms of the FFT. Recall that the FFT essentially computes the multiplication of the Vandermonde matrix with the coefficients to get to the outputs, e.g. $V\vec{a} = \vec{y}$.

To go from the outputs to the coefficients, we can simply

multiply by V^{-1} , i.e. $\vec{a} = V^{-1}\vec{v}$.

FF1

Huar

Introduction

Algorithm

Naive FTT

Point-Value Complex Roots

NTT

Application

2D Convolution

onclusion

Sample Problem

Past Lectures

References

Computing V^{-1} is tedious and I don't have much insight (read *Introduction to Algorithms* for a proper proof), but it essentially involves just the definition of matrix inverse and more properties of roots of unity.

FF1

Huar

Introduction

Algorithm

FTT
Point-Value
Complex Roots

Iterative

Applications

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Computing V^{-1} is tedious and I don't have much insight (read *Introduction to Algorithms* for a proper proof), but it essentially involves just the definition of matrix inverse and more properties of roots of unity.

It turns out that V^{-1} is essentially V but evaluated at x^{-1} instead of x. Also, divide by n.

FF"

Huar

Introductio

Algorithm Naive

Point-Value
Complex Roots
Iterative

Application

2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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It turns out that V^{-1} is essentially V but evaluated at x^{-1} instead of x. Also, divide by n.

So we can just use the FFT but take the inverse of the root of unity, and divide each element by n at the end.

FF"

Huar

Introductio

Algorithm Naive

Point-Value
Complex Roots
Iterative

Application

2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

Computing V^{-1} is tedious and I don't have much insight (read *Introduction to Algorithms* for a proper proof), but it essentially involves just the definition of matrix inverse and more properties of roots of unity.

It turns out that V^{-1} is essentially V but evaluated at x^{-1} instead of x. Also, divide by n.

So we can just use the FFT but take the inverse of the root of unity, and divide each element by n at the end.

Finally, we arrive at the FFT formulation of convolutions.

Convolution Theorem

FF1

Huar

Introductio

......

Naive

Point-Value

Complex Roots

NTT

Application

2D Convolution: Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

Theorem

 $f * g = FFT^{-1}(FFT(f) \circ FFT(g))$, i.e. convolutions can be done with FFTs in time $O(n \log n)$.

Proof.

Follows from the presentation up to this point.

A concrete implementation can be found here.

Iterative FFT

FFT

Huar

Introductio

Algorithms

Naive

Point-Value Complex Roo

Iterative

Application

2D Convolutions Separable Kerne

onclusion

Sample Problem

Past Lecture

References

The recursive algorithm can be made iterative surprisingly elegantly from a pattern in binary form of the indexes when recursively subdividing.

Iterative FFT

FF1

Huar

Introductio

Algorithm

Naive

Point-Value Complex Roo

Iterative NTT

Applications

Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

The recursive algorithm can be made iterative surprisingly elegantly from a pattern in binary form of the indexes when recursively subdividing.

I omit the details here, although it makes the algorithm O(n) in memory instead of $O(n \log n)$ and will likely run faster than the recursive algorithm.

Iterative FFT

FF1

Huar

Introductio

Algorithms

Naive FTT

Point-Value
Complex Root

Application

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

The recursive algorithm can be made iterative surprisingly elegantly from a pattern in binary form of the indexes when recursively subdividing.

I omit the details here, although it makes the algorithm O(n) in memory instead of $O(n \log n)$ and will likely run faster than the recursive algorithm.

An implementation is above.

Number Theoretic Transform (NTT)

FF1

Huar

Introductio

Algorithma

Naiv

Point-Value Complex Roo

NTT

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lectures

References

Another improvement on the FFT comes from the observation that complex roots of unity were an arbitrary pick, any field with sufficient properties will do.

Number Theoretic Transform (NTT)

FF1

Huar

Introductio

Naiv

NTT

Point-Value Complex Roo Iterative

Application

Audio
2D Convolutions
Separable Kernel

Conclusion

Sample Problem

Past Lecture

References

Another improvement on the FFT comes from the observation that complex roots of unity were an arbitrary pick, any field with sufficient properties will do.

In particular, we can pick a large prime number p and find an equivalent to a root of unity under the field modulo p.

Uses of the NTT

FF1

Huar

Introductio

Algorithm

Naive

Point-Value Complex Roo

NTT

Application:

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

The details are incredibly tedious and number theory heavy, but they yield the number theoretic transform, a variant of the FFT which operates on integers.

Use cases of the NTT

The NTT is useful for polynomials of integer coefficients or certain types of data, e.g. music or images, which have integer pixel values.

Accounting for Negative Numbers

FFT

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root

NTT

Application

2D Convolutions Separable Kernel

Conclusion

Sample Problem

Past Lectures

Deferences

One downside is that negative numbers do not exist under modulo, which can be accounted for by assuming large numbers are in fact negative, changing the range from [0,p) to $[-\frac{p}{2},\frac{p}{2})$.

```
def ntt_sign(1: list, p: int) -> list:
  return [x if x < (p >> 1) else x - p for x in 1]
```

Table of Contents

FFT

Huan

Introduction

Naive FTT

> Point-Value Complex Roots

ΝΤΊ

Application

Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

- 1 Introduction
 - 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- Doot Looking
- 7 References

The Anime Music Quiz Problem

FF1

Huar

Introduction

Algorithm

Naive

Point-Value
Complex Root
Iterative

Applications

Audio

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

Before we apply 2D convolutions to images, we elucidate the 1D convolution and its usefulness through an illustrative example.

The anime music quiz problem.

We have a song that is 1 minute and 30 seconds long, and a 10 second clip from that song. We wish to compute:

The Anime Music Quiz Problem

FF1

Huar

Introduction

Algorithm

Naive

Point-Value
Complex Root
Iterative

Applications

Audio 2D Convolution Separable Kerne

Conclusion

Sample Problem

Past Lectures

References

Before we apply 2D convolutions to images, we elucidate the 1D convolution and its usefulness through an illustrative example.

The anime music quiz problem.

We have a song that is 1 minute and 30 seconds long, and a 10 second clip from that song. We wish to compute:

1 Out of a list of songs, which song the clip came from.

The Anime Music Quiz Problem

FF1

Huar

Introduction

Algorithm

Naive FTT

Point-Value Complex Root Iterative

Applications

Audio 2D Convolutio

Separable Kerne
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

Before we apply 2D convolutions to images, we elucidate the 1D convolution and its usefulness through an illustrative example.

The anime music quiz problem.

We have a song that is 1 minute and 30 seconds long, and a 10 second clip from that song. We wish to compute:

- 1 Out of a list of songs, which song the clip came from.
- From a known song, the timestamp where the clip occurred.

Example Run

FF1

Huar

Introduction

Algorithms

Naive

Point-Value Complex Roots Iterative

Applications

2D Convolutions
Separable Kernel

Conclusion

Sample Problems

Past Lectures

References

```
(amq) stephenhuan@MacBook-Pro ~/P/p/p/amq (master o)> python db.py clip Picking a clip from NeonGenesisEvangelion at -4.04dB
```

(a) Generating a clip from an anime intro.

NekomonogatariKuro-OP1 : 4590137 loss, occurs at 3.4 seconds in NeonGenesisEvangelion : 3931757 loss, occurs at 35.2 seconds in Nichijou op2 : 8316782 loss, occurs at 42.4 seconds in Nichijou op2 | Nisemonogatari_op1 : 4663729 loss, occurs at 77.3 seconds in

(b) Comparison between songs; finds it occurs exactly 35.2 seconds in

Final answer: Neon Genesis Evangelion 123 songs in 8.61 seconds = 14.29 songs per second

(c) Song with the lowest loss.

Figure: An example run of the system.

EF1

Huar

Introduction

Algorithn

Naive

Point-Value Complex Ro

NTT

Application Audio

2D Convolutions Separable Kerne

onclusion

Sample Problems

Past Lectures

References

First, some basics about the representation of audio data.

EE1

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Application

Audio

O Convolutions parable Kernel FT Algorithm

onclusion

Sample Problem

Past Lecture

References

First, some basics about the representation of audio data. We will use the mp3 file format at a sample rate of 48kHz.

FF1

Huar

Introduction

Algorithm

FTT

Complex Root
Iterative

Application

2D Convolution

Conclusion

Sample Problem

Past Lectures

References

First, some basics about the representation of audio data. We will use the mp3 file format at a sample rate of 48kHz. Audio is fundamentally just a list of numbers, where each number represents the amplitude of the sound wave at that time. A 48kHz sample rate means there are 48,000 of these measurements per second.

FF"

Huar

Introduction

Algorithm

Point-Value Complex Roc

Complex Roof
Iterative
NTT

Application Audio

2D Convolutions Separable Kernels FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

First, some basics about the representation of audio data. We will use the mp3 file format at a sample rate of 48kHz. Audio is fundamentally just a list of numbers, where each number represents the amplitude of the sound wave at that time. A 48kHz sample rate means there are 48,000 of these measurements per second.

Each number is a 16-bit number in the range [0,1), which we will transform to an integer in the range $[0,2^{16})$ for the NTT.

Modeling the Problem

FF1

Huar

Introduction

Naive

Point-Value
Complex Roots
Iterative

Application

Audio

D Convolutions eparable Kernel FT Algorithm

onclusion

Sample Problems

Past Lectures

References

So we have two lists of integers, and now wish to find where the smaller list "fits" into the larger list the best.

Modeling the Problem

FF1

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root

nnlication

Application

2D Convolution
Separable Kerne
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

So we have two lists of integers, and now wish to find where the smaller list "fits" into the larger list the best.

One way to do this is to compute the ℓ^2 norm, or the vector difference between the two lists.

Modeling the Problem

FF1

Huar

Introductio

Algorithms
Naive
FTT
Point-Value
Complex Root
Iterative

Applications Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

So we have two lists of integers, and now wish to find where the smaller list "fits" into the larger list the best.

One way to do this is to compute the ℓ^2 norm, or the vector difference between the two lists.

So we slide the smaller list over the larger list, computing the sum of squares error as we go. Note that this is very similar to the convolution, except we to calculate the sum of squares instead of the element-wise product. We also need to flip one of the lists because the convolution flips a list.

Turning Sum of Squares into Convolutions

FFT

Huai

Introductio

Naive

Point-Value Complex Roo

Iterative

NTT

Application

Audio

D Convolutions eparable Kernel FT Algorithm

onclusion

Sample Problem

Past Lectures

References

How do we reduce sum of squares to an element-wise product?

Turning Sum of Squares into Convolutions

FFT

Huar

Introduction

A I 24 I

Naive

FTT Point-

Complex Roots

Application

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

How do we reduce sum of squares to an element-wise product? We notice that for elements of the lists a, b

$$(a_i - b_j)^2 = a_i^2 - 2a_ib_j + b_j^2$$

Turning Sum of Squares into Convolutions

FFT

Huar

Introduction

Algorithms

Naive FTT

Point-Value
Complex Root
Iterative

Application

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

How do we reduce sum of squares to an element-wise product? We notice that for elements of the lists a, b

$$(a_i - b_j)^2 = a_i^2 - 2a_ib_j + b_j^2$$

When we sum over the length of a, assuming a is the smaller list, we get:

$$||a||^2 - 2a \cdot b' + ||b'||^2$$

where b' is the slice that a overlaps.

Turning Sum of Squares into Convolutions

FFT

Huar

Introduction

Algorithms

FTT
Point-Value

Complex Root
Iterative

Application:

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

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When we sum over the length of a, assuming a is the smaller list, we get:

$$||a||^2 - 2a \cdot b' + ||b'||^2$$

where b' is the slice that a overlaps. ||a|| is a constant, so it can be ignored. Thus, we only need to compute $a \cdot b'$ and ||b'||.

Turning Sum of Squares into Convolutions

FFT

Huar

Introduction

Algorithms

FTT
Point-Value
Complex Roo

Complex Root
Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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When we sum over the length of a, assuming a is the smaller list, we get:

$$||a||^2 - 2a \cdot b' + ||b'||^2$$

where b' is the slice that a overlaps. ||a|| is a constant, so it can be ignored. Thus, we only need to compute $a \cdot b'$ and ||b'||. $a \cdot b'$ directly follows from a convolution and can be read from a * b', where b' is the reverse of b.

Turning Sum of Squares into Convolutions

FFT

пиап

Introductio

Algorithms

Naive FTT

Point-Value
Complex Root
Iterative

Application: Audio

Separable Kernel

Conclusion

Sample Problems

Past Lectures

Referen

How do we reduce sum of squares to an element-wise product? We notice that for elements of the lists a, b

$$(a_i - b_j)^2 = a_i^2 - 2a_ib_j + b_j^2$$

When we sum over the length of a, assuming a is the smaller list, we get:

$$||a||^2 - 2a \cdot b' + ||b'||^2$$

where b' is the slice that a overlaps. $\|a\|$ is a constant, so it can be ignored. Thus, we only need to compute $a \cdot b'$ and $\|b'\|$. $a \cdot b'$ directly follows from a convolution and can be read from $a * b^r$, where b^r is the reverse of b. Lastly, if we make sure to scan from left to right, then we can compute $\|b'\|$ by storing an intermediate value, and updating it when we slide a an additional index by subtracting out the front of b', where a left from, and adding the new value that a covers.

FF"

Huar

Introduction

Algorithms

Naive

Point-Value Complex Roots Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

onclusion

Sample Problems

Past Lectures

References

Algorithm minimum ℓ^2 between two lists

```
def min_offset(a: list, b: list) -> tuple:
    N, M = len(a), len(b)
    p = fft(a[::-1], b)[N - 1:]
    x2, xy, y2 = sum(x*x for x in a), <math>p[0], \
                 sum(b[i]*b[i] for i in range(N))
    best, 12 = 0, -2*xy + y2
    for i in range(1, M - N + 1):
        y2 += b[N - 1 + i]*b[N - 1 + i] - b[i - 1]*b[i - 1]
        xy = p[i]
        d = -2*xy + y2
        if d < 12:
            best. 12 = i.d
    return best, x2 + 12
```

FFT

Huai

Introduction

Naive

Point-Value Complex Root

Iterative NTT

Application

Audio

D Convolutions sparable Kerne FT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We need to be careful about a few things. If we don't pick p for the NTT large enough, then it won't work.

FF1

Huai

Introduction

Algorithms

Naive FTT

Point-Value
Complex Roots
Iterative

 ${\sf Application}$

Audio

2D Convolution Separable Kerne FFT Algorithm

onclusion

Sample Problem

Past Lectures

References

We need to be careful about a few things. If we don't pick p for the NTT large enough, then it won't work.

If m is the largest number in a list and n is the length of the list, then we need p to be bigger than m^2n , the largest a single element can become.

 $m^2 n = 2^{32} \cdot 4 \cdot 10^6 \approx 2^{54}$

FF1

Huai

Introduction

Algorithm Naive

Point-Value
Complex Root

NTT

Application

Audio

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We need to be careful about a few things. If we don't pick p for the NTT large enough, then it won't work. If m is the largest number in a list and n is the length of the list, then we need p to be bigger than m^2n , the largest a single element can become. n is $90 \cdot 48,000 \approx 4 \cdot 10^6$ and m is 2^{16} .

FF1

Huai

Introduction

Naive FTT Point-Value

Point-Value Complex Root Iterative

Applications
Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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n is $90 \cdot 48,000 \approx 4 \cdot 10^6$ and m is 2^{16} .

 $m^2 n = 2^{32} \cdot 4 \cdot 10^6 \approx 2^{54}.$

This seems fine since 2^{54} will fit in a long, but this won't work since we need to compute x^2 as part of the FFT, and $(2^{54})^2$ will definitely overflow.

FF1

Huai

Introduction

Algorithms
Naive
FTT
Point-Value
Complex Roo
Iterative

Applications Audio 2D Convolution

2D Convolution Separable Kerne FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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n is $90 \cdot 48,000 \approx 4 \cdot 10^6$ and m is 2^{16} .

$$m^2 n = 2^{32} \cdot 4 \cdot 10^6 \approx 2^{54}.$$

This seems fine since 2^{54} will fit in a long, but this won't work since we need to compute x^2 as part of the FFT, and $(2^{54})^2$ will definitely overflow.

We could get around this overflow by doing modulo multiplication instead of standard multiplication, but that would introduce a log factor, making the algorithm 64x slower, an unacceptable slowdown.

FFT

Huar

Introduction

Algorithm

Naive

Point-Value
Complex Root

Audio

D Convolutions Separable Kernel FFT Algorithm

onclusion

Sample Problem

Past Lectures

References

One trick is to reduce the bitrate of the mp3 at the sacrifice of audio quality, and go from 16-bit audio to 8-bit audio.

FFT

Huar

Introduction

Algorithm Naive

Point-Value Complex Root Iterative

Applications

Audio

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

One trick is to reduce the bitrate of the mp3 at the sacrifice of audio quality, and go from 16-bit audio to 8-bit audio.

A naive way to do it would be to multiply the real number by 2^8 and round, but a better way is the μ -law algorithm, a trick that preserves frequencies closer to the human voice. A comparison between naive scaling and the μ law is presented here.

FF1

Huar

Introductio

Algorithm
Naive

Point-Value Complex Roo Iterative NTT

Applications

Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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A naive way to do it would be to multiply the real number by 2^8 and round, but a better way is the μ -law algorithm, a trick that preserves frequencies closer to the human voice. A comparison between naive scaling and the μ law is presented here.

With 8-bit audio, $m^2n = 2^{16} \cdot 4 \cdot 10^6 \approx 2^{38}$. This goes over the limit of 2^{32} for x^2 to fit in a long, but it works in practice since mp3 audio rarely hits maximum volume and our clip is 10

seconds long, and we computed for 90 seconds.

FF1

Huai

Introduction

Algorithm Naive FTT

Point-Value Complex Roo Iterative NTT

Applications Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

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With 8-bit audio, $m^2n=2^{16}\cdot 4\cdot 10^6\approx 2^{38}$. This goes over the limit of 2^{32} for x^2 to fit in a long, but it works in practice since mp3 audio rarely hits maximum volume and our clip is 10 seconds long, and we computed for 90 seconds.

An implementation is given here and a video walkthrough here.

Table of Contents

FFT

Huar

Introduction

A.1. 1.1

Naive

Point-Value Complex Root

итт

Application:

2D Convolutions

FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

- 1 Introduction
 - 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- 5 Dample Froblems
- 7 References

2D Convolutions

FF1

Huai

Introduction

Algorithm

Point-Value Complex Root Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

2D convolutions, a convolution generalized to matrices, are useful in computer vision for a variety of reasons, including edge detection and convolutional neural networks. Their exact usage will not be discussed here, and instead we will discuss an efficient way to calculate a 2D convolution with the FFT we have already developed.

Definitions

We have an "data" matrix, representing an image, and we have a *kernel* matrix, which is the matrix we imagine sliding over the image. This is also known as a *filter*.

Scipy Definition

FF1

Huar

Introductio

Algorithms

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

For 2D convolutions, the result is slightly ambiguous depending on how one defines it.

We will use scipy's definition, where to calculate the value of the convolution at a particular point, we imagine the *bottom right* corner of the kernel placed over that point.

Convolution Example

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Roots Iterative

Application

2D Convolutions Separable Kernel

Conclusior

Sample Problem

Past Lecture

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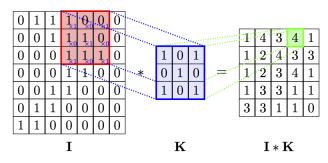


Figure: A convolution taken from here.

Definition of a 2D Convolution

FF1

Huar

Introduction

Algorithms

Naive FTT

Point-Value Complex Root Iterative

Application

2D Convolutions Separable Kernel

Conclusion

Sample Problem

Past Lectures

References

We define the 2D convolution between an image x of size MxN and a kernel h of size HxW as follows (similar to the 1D case, we assume both matrices are padded with 0's):

Definition of the 2D convolution

$$(x*h)[i,j] = \sum_{k=0}^{i} \sum_{l=0}^{j} x[k][l]h[i-k][j-l]$$

FF1

Huar

.....

Naive

Point-Value
Complex Root

Applications

Audio
2D Convolutions

Separable Kerne

Conclusion

Sample Problem

Past Lectures

References

This operation is also symmetric, so what we call the image and the kernel is essentially arbitrary (by convention, the kernel is the smaller matrix). The resulting matrix is going to be of size $(M+H-1)\times(N+W-1)$ from the same logic as the 1D case.

FF"

Huai

Introductio

Algorithms

Naive FTT

Complex Root
Iterative

Applications

2D Convolutions
Separable Kernel

Conclusion

Sample Problem

Past Lecture

References

This operation is also symmetric, so what we call the image and the kernel is essentially arbitrary (by convention, the kernel is the smaller matrix). The resulting matrix is going to be of size $(M+H-1)\mathbf{x}(N+W-1)$ from the same logic as the 1D case. Thus, the time it takes to compute the convolution is O(MNHW). We can, however, take advantage of a trick if the kernel has a certain property.

Table of Contents

FFT

Huan

Introduction

Algorithms

Naive

Point-Value Complex Roots

NTT

Application

2D Convolutions
Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

- 1 Introduction
- 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform
- 3 Applications of Convolutions
 - Audio Processing
 - 2D Convolutions
 - Separable Kernels
 - FFT Algorithm
 - Conclusion
- 5 Sample Problems
- 5 Sample i Toblems
- 7 References

Separable Kernels

FFT

Huar

Introduction

Algorithms

Naive

Point-Value Complex Root

Iterative

Applicatior

Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

Separable

A matrix M is **separable** if it can be written as $\vec{u}\vec{v}^T$ for some vectors \vec{u}, \vec{v} .

Separable Kernels

FFT

Huar

Introduction

Algorithm

Naive FTT

> Point-Value Complex Roots Iterative

Application

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

Separable

A matrix M is **separable** if it can be written as $\vec{u}\vec{v}^T$ for some vectors \vec{u} , \vec{v} .

Sobel matrix

The famous Sobel matrix for edge detection is separable:

$$\begin{bmatrix} 1 & 0 & -1 \\ 2 & 0 & -2 \\ 1 & 0 & -1 \end{bmatrix} = \begin{bmatrix} 1 \\ 2 \\ 1 \end{bmatrix} \begin{bmatrix} 1 & 0 & -1 \end{bmatrix}$$

Separable Kernels

FFT

Huar

Introduction

Algorithm Naive

Point-Value

Complex Root

Applications

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Separable

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Separating the convolution

If M is separable, we can convolute with \vec{u} and then with \vec{v} .

Proof of Separability

FFT

Huar

Introduction

Algorithms

Naive FTT

Point-Value
Complex Root
Iterative

Application Audio

Separable Kernels
FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Theorem

If $h = \vec{u}\vec{v}^T$, then $(x * h) = ((x * \vec{u}) * \vec{v})$, i.e. we can separate a convolution into two parts.

Proof.

$$(x * u)[i,j] = \sum_{k=0}^{i} \sum_{l=0}^{j} x[k][l]u[i-k][j-l]$$
 Definition

Since u is a column vector, it only has values when l = j, removing the inner sum.

$$= \sum_{k=0}^{i} x[k][j]u[i-k][0]$$

Proof of Separability

FFT

Huai

Introductio

Algorithms

Naiv

Point-Value Complex Root Iterative

Applications Audio

Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

Proof.

Convoluting with v,

$$((x*u)*v)[i,j] = \sum_{k=0}^{i} \sum_{l=0}^{j} (\sum_{m=0}^{k} x[m][l]u[k-m][0])v[i-k][j-l]$$

Since v is a row vector, it only has values when k = i, removing the outermost sum.

$$=\sum_{l=0}^{j}(\sum_{m=0}^{i}\times[m][l]u[i-m][0])v[0][j-l]$$

Proof of Separability

FFT

Huan

Introduction

Algorithms

Naive

Point-Value

Complex Roots
Iterative

Application

2D Convolutions
Separable Kernels

Conclusion

Sample Problems

Past Lectures

References

Proof.

Swapping the order of summations and renaming m to k,

$$= \sum_{k=0}^{i} \sum_{l=0}^{j} x[k][l]u[i-k][0]v[0][j-l]$$

From the fact that h[x][y] = u[x][0]v[0][y],

$$= \sum_{k=0}^{i} \sum_{l=0}^{j} x[k][l]h[i-k][j-l]$$

= (x * h)

Runtime Analysis

FFT

Huar

Algorithms

Naive

Point-Value

Iterativ

Application

2D Convolutions

Separable Kernels FFT Algorithm

onclusion

Sample Problems

Past Lectures

References

How does this help us?

Runtime Analysis

FF1

Huar

Introductio

Algorithms

Naive

Point-Value

Iterative

Applicatio

Audio

Separable Kernels

Conclusion

Sample Problem

Past Lecture

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How does this help us?

Well, recall the running time of O(MNHW). If we do two convolutions of a kernel of Hx1 and another of 1xW, the running time will be O(MNH + MNW) = O(MN(H + W)), a significant improvement as HW grows quadratically while H + W grows linearly.

Runtime Analysis

FF1

Huar

Introductio

Naive
FTT
Point-Value
Complex Root
Iterative

Applications

2D Convolutions Separable Kernels FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

How does this help us?

Well, recall the running time of O(MNHW). If we do two convolutions of a kernel of Hx1 and another of 1xW, the running time will be O(MNH + MNW) = O(MN(H + W)), a significant improvement as HW grows quadratically while H + W grows linearly.

We can also use repeated 1D convolution to compute the 2D convolution for the specific case of a vector, yielding a $O(MN \log MN)$ time algorithm.

FF1

Huar

miroductic

Algorithm

Naive FTT

Point-Value
Complex Root
Iterative

Applications Audio

Separable Kernels
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

However, not every matrix is separable. The conditions are quite strict, a matrix is separable if and only if every pair of rows is a multiple of each other, i.e. the matrix is made up of multiples of a particular row vector. As a consequence, the matrix is also made up of multiples of a particular column vector. These matrices are relatively rare, so there is utility in deriving a more general algorithm.

Table of Contents

FFT

Huan

Introduction

Algorithms

Naive

Point-Value Complex Roots

итт

Application:

2D Convolutions
Separable Kernels

FFT Algorithm

Conclusion

Sample Problems

Past Lectures

Referenc

- 1 Introduction
- 2 Algorithms
 - Naive
 - Fast Fourier Transform
 - Point-Value Representation
 - Complex Roots of Unity
 - Iterative Variant
 - Number Theoretic Transform

3 Applications of Convolutions

- Audio Processing
- 2D Convolutions
- Separable Kernels
- FFT Algorithm
- 4 Conclusion
- 5 Sample Problems
- Doot Looking
- 7 References

Reduction to 1D Convolution

FFT

Huar

Introduction

Algorithm

Point-Value

Complex Roots
Iterative
NTT

Application

Audio
2D Convolutions
Separable Kerne
FFT Algorithm

onclusion

Sample Problem

Past Lectures

References

We can reduce 2D convolutions to 1D convolutions if we're clever. The observation is that if we flatten both matrices into a 1D list by reading from top to bottom, left to right, we can just convolute in 1D and reconstruct the matrix afterwards.

Reduction to 1D Convolution

FFT

Huar

Introductio

Algorithm

Point-Value
Complex Root
Iterative

Applications
Audio

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

We can reduce 2D convolutions to 1D convolutions if we're clever. The observation is that if we flatten both matrices into a 1D list by reading from top to bottom, left to right, we can just convolute in 1D and reconstruct the matrix afterwards. We need to make sure both matrices are sufficiently padded with zeros, such that the zeros force values in the kernel to their proper rows in the image.

Reduction to 1D Convolution

FF1

Huan

Introductio

Algorithm
Naive
FTT

Point-Value
Complex Roo
Iterative
NTT

Applications
Audio
2D Convolutions
Separable Kernel

Conclusion

Sample Problems

Past Lectures

References

We can reduce 2D convolutions to 1D convolutions if we're clever. The observation is that if we flatten both matrices into a 1D list by reading from top to bottom, left to right, we can just convolute in 1D and reconstruct the matrix afterwards. We need to make sure both matrices are sufficiently padded with zeros, such that the zeros force values in the kernel to their proper rows in the image.

It turns out that we can just pad both matrices to the final column size of the convolution, N+W-1, flatten both, convolute with the FFT, and then reshape the resulting list to a matrix of proper size.

Summary

FF1

Huar

Introductio

Algorithms

FTT Point

Point-Value
Complex Roots
Iterative

Application

Audio
2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

lacktriangle Pad the rows of both matrices with zeros such that each has a width of N+W-1

Summary

FFT

Huar

Introductio

Algorithn Naive

> Point-Value Complex Roots Iterative

Application

Audio
2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

Deferences

- $lue{1}$ Pad the rows of both matrices with zeros such that each has a width of N+W-1
- 2 Flatten both by reading top to bottom, left to right

Summary

FF"

Huar

Introductio

Naive FTT Point-Valu Complex R

Point-Value
Complex Roots
Iterative
NTT

Application:

2D Convolutions
Separable Kernel

onclusion

Sample Problem

Past Lectures

References

- \blacksquare Pad the rows of both matrices with zeros such that each has a width of N+W-1
- 2 Flatten both by reading top to bottom, left to right
- 3 Convolute the resulting lists in 1D

Summary

FF1

Huar

Introductio

Naive
FTT
Point-Value
Complex Roots
Iterative

Application:

2D Convolutions
Separable Kernels

Conclusion

Sample Problem

Past Lectures

References

- 1 Pad the rows of both matrices with zeros such that each has a width of N+W-1
- 2 Flatten both by reading top to bottom, left to right
- 3 Convolute the resulting lists in 1D
- 4 Reconstruct a 2D matrix, because we know its shape is (M+H-1)x(N+W-1)

The 2D Convolution Algorithm

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FF1
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Huar

Introduction

Algorithm

FTT
Point-Value
Complex Root

Applications

2D Convolutions Separable Kernel FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

Algorithm 2D Convolution Algorithm

```
def flatten(m: list, pad=0) -> list:
    """ Flattens a matrix into a list.
   return [x for row in m for x in row + [0]*pad]
def reshape(1: list, m: int, n: int) -> list:
    """ Shapes a list into a MxN matrix."""
   return [[l[r*n + c] for c in range(n)]
            for r in range(m)]
def conv(h: list, x: list):
    """ Computes the 2D convolution. """
    M, N, H, W = len(x), len(x[0]), len(h), len(h[0])
    # need to pad the columns to the final size
   h, x = flatten(h, N - 1), flatten(x, W - 1)
   return reshape(fft(h, x), M + H - 1, N + W - 1)
```

FF1

Huar

Introduction

Algorithm

Naive

Point-Value Complex Roots Iterative

Application

2D Convolutions
Separable Kerne

Conclusion

Sample Problem

Past Lectures

References

In most computer vision applications, the kernel is a square matrix of size $K \times K$, where K is an odd number.

FF

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Applications

2D Convolutions Separable Kerne

Conclusion

Sample Problem

Past Lecture

Poforonco

In most computer vision applications, the kernel is a square matrix of size $K \times K$, where K is an odd number.

The *middle value* of the kernel is then placed over each pixel of the image, yielding a transformed image of the same dimensionality as the original.

FF"

Huar

Introductio

Algorithn Naive

> Point-Value Complex Root Iterative

Applications
Audio
2D Convolutions
Separable Kernel
FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

In most computer vision applications, the kernel is a square matrix of size $K \times K$, where K is an odd number.

The *middle value* of the kernel is then placed over each pixel of the image, yielding a transformed image of the same dimensionality as the original.

We can simulate this by simply cutting off the first and last $\frac{K-1}{2}$ rows and the same for the columns. This transforms the resulting size from N+K-1 to $N+K-1-2\frac{K-1}{2}=N$.

FF1

Huai

Introductio

```
Algorithm:
```

Naive

Point-Value
Complex Roots
Iterative

Applications

2D Convolutions
Separable Kernel

FFT Algorithm

Conclusion

Sample Problem

Past Lecture

Poforonco

```
_____ pruning _______

def prune(h: list, x: list) -> list:
    """ Prunes a convolution for the specific KxK
    → filter case. """
    m, k = conv(h, x), min(len(h), len(x))
    pad = (k - 1)>>1
    return [row[pad:-pad] for row in m[pad:-pad]]
```

Runtime Analysis

FF1

Huar

Introduction

Algorithms

Naive

Point-Value
Complex Roots
Iterative

Application .

2D Convolutions
Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

The running time of the algorithm is going to be $O(MN \log MN) = O(MN(\log N + \log M) = O(MN \log N)$ since we convolute a list of length M(N + W - 1), and we assume N > M > W.

Runtime Analysis

FF1

Huar

Introduction

Algorithms

Point-Value
Complex Root
Iterative
NTT

Applications

Audio

Separable Kerne

Conclusion

Sample Problem

Past Lecture

References

The running time of the algorithm is going to be $O(MN \log MN) = O(MN(\log N + \log M) = O(MN \log N)$ since we convolute a list of length M(N + W - 1), and we assume $N \ge M > W$.

This is not necessarily faster than the brute-force algorithm; it depends on the kernel size. For simplicity, suppose we have a $N \times N$ image and a $K \times K$ kernel where N > K. Brute force yields $O(N^2 K^2)$ while the FFT algorithm yields $O(N^2 \log N)$.

Runtime Analysis

FF"

Huar

Introduction

Algorithm
Naive
FTT

Complex Roo Iterative

Audio
2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

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This is not necessarily faster than the brute-force algorithm; it depends on the kernel size. For simplicity, suppose we have a $N \times N$ image and a $K \times K$ kernel where N > K. Brute force yields $O(N^2 K^2)$ while the FFT algorithm yields $O(N^2 \log N)$. Thus, if $\log N < K^2$ then the FFT algorithm is going to be faster. For K > 5 that is a fair assumption since $K^2 = 25$, 2^{25} is several million. Obviously the FFT algorithm has a much larger constant factor, but for a sufficiently large kernel the time savings become greater and greater.

FF1

Huar

Introductio

Algorithm

Naive FTT

Point-Value Complex Root Iterative

Application

2D Convolutions Separable Kerne

Conclusion

Sample Problems

Past Lectures

References

The convolution, an operator very useful for signal, audio, and image processing, can be efficiently computed with the Fast Fourier Transform, or FFT.

FF1

Huar

Introductio

Algorithm

Naive

Point-Value Complex Root Iterative

Application:

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problem

Past Lecture

References

The convolution, an operator very useful for signal, audio, and image processing, can be efficiently computed with the Fast Fourier Transform, or FFT.

If the data is integer, then floating-point arithmetic can be avoided with the Number Theoretic Transform (NTT), a variant of the FFT which uses modulo instead of complex numbers, and calculates entirely in integers.

FF1

Huar

Introduction

Algorithms

Naive

Point-Value Complex Roots Iterative

Application

2D Convolution

Conclusion

Sample Problem

Past Lecture

References

This lecture skips over the continuous case (what I've been calling the Fast Fourier Transform is more mathematically called the Discrete Fourier Transform, or DFT) but the idea is essentially the same, any summation turns into an integral.

FF1

Huar

Introduction

Algorithms

Naive FTT

Point-Value
Complex Roo
Iterative

Audio

2D Convolutions

Separable Kernel

Conclusion

Sample Problems

Past Lectures

References

This lecture skips over the continuous case (what I've been calling the Fast Fourier Transform is more mathematically called the Discrete Fourier Transform, or DFT) but the idea is essentially the same, any summation turns into an integral. It also skips over the mathematical interpretation of the FFT, involving decomposing a function into a series of sine and cosine waves. This is useful for signal processing and audio analysis, but requires a stronger mathematical background and to be honest, I haven't studied it at all myself. Fourier analysis goes deeper than we need here.

FF1

Huai

Introductio

Algorithm

Naive

Point-Value Complex Root

NTT

Application: Audio

2D Convolutions
Separable Kernels
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

Introduction to Algorithms is definitely the most helpful source on the FFT (from a computer science perspective), and more thorough treatments of the FFT from an engineering or mathematical standpoint are not hard to find.

Examples of problems using the FFT

FFT

Huan

Introductio

Algorithms

Naive

Point-Value Complex Roots

Iterative

Application

Audio 2D Convolutions Separable Kernel

onclusion

Sample Problems

Past

Deferences

1 SPOJ POLYMUL: Direct application of the FFT.

Examples of problems using the FFT

Sample **Problems**

2 SPOJ MUL: Given 1000 pairs of numbers, compute the product of each pair; each number can have up to 10,000 digits.

Examples of problems using the FFT

FFT

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Applications

2D Convolutions
Separable Kerne

Conclusion

Sample Problems

Past Lectures

References

2 Solution: Think of numbers as polynomials, where the digits are coefficients and x is 10. Then, you can multiply two numbers by multiplying the polynomials. However, there is no guarantee that the coefficients of the resulting polynomial are less than 10, so it is not a valid number. As a last post-processing step, start from the smallest place value and move your way to the largest, moving the digit overflow from one place value to the next. Since you iterate over the number of digits in the number, it takes $O(\log n)$ which is dominated by the FFT.

Examples of problems using the FFT

FFT

Huar

Introductio

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions
Separable Kerne

Conclusion

Sample Problems

Past Lecture

Deferences

2 An extension of this idea is the Schönhage-Strassen algorithm, which disregards the requirement that the intermediate numbers fit in a long, at the cost of being $O(n \log n \log \log n)$. A more recent algorithm, by Harvey and van der Hoeven, achieves $O(n \log n)$.

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

References

3 SPOJ MAXMATCH: Given a string S of length N made up of the characters "a", "b", and "c", compute the maximum self-matching, where a self-matching is defined as the number of characters which match between S and S shifted some nonzero number of characters.

Examples of problems using the FFT

FFT

Huar

Introduction

Algorithm

Naive

Point-Value
Complex Root
Iterative

Application

Audio

2D Convolutions

Separable Kernel

FFT Algorithm

Conclusion

Sample Problems

Past Lectures

References

3 Solution: For an offset i, the size of the overlap will be N-i. So we just need to find the number of differences, and subtract that from N-i to obtain the number of matches. The easiest thing to do is to keep track of each character separately, so to compute the differences for each character. Suppose our character is "a". We encode "a" as a 0, and the other characters as a 1. We then find the ℓ^2 norm between this new list and this list with N 1's added to it (so that when we overlap, the non "a" characters aren't counted). This has the complication of counting "a"'s which are off the edge of the string, which we can account for by simply keeping track of the number of "a"'s we have seen.

Examples of problems using the FFT

FFT

Huar

Introductio

Naive

Point-Value Complex Roots Iterative

Application

2D Convolution Separable Kerne

onclusion

Sample Problems

Past Lecture

Deferences

3 Given a[i] as the number of mismatches with the character "a" at a shift of i, and b[i], c[i], the number of matches is $N-i-\frac{a[i]+b[i]+c[i]}{2}$. We divide by 2 because we count each mismatch twice (once for each character in the pair).

Examples of problems using the FFT

FFT

Huai

Introduction

.

Naive FTT Point-Value Complex Root

Application

2D Convolutions
Separable Kerne
FFT Algorithm

Conclusion

Sample Problems

Past Lecture

References

3 A much conceptually simpler algorithm is to encode "a", "b", and "c" cleverly and then compute the matches in one shot. If we encode "a" as (1, 0, 0), "b" as (0, 1, 0), and "c" as (0, 0, 1), the FFT of the resulting list with its reverse will give us the number of matches at each index because the character representations dot each other will be 1 if they are equal, and 0 if they are unequal. Thus, the FFT will give us exactly the number of matches, but we need to only look at every 3rd index since the other 2 are byproducts of our transformation.

In practice, running one big FFT is faster than running 3 smaller FFTs

Examples of problems using the FFT

FFT

Huar

Introductio

A I * L

Naive

Point-Value Complex Root

Iterative

Application

2D Convolution

onclusion

Sample Problems

Past Lecture

Deferences

4 Codechef FARASA: Given an array, find the number of distinct sums of a contiguous subarray.

Examples of problems using the FFT

FFT

Huar

Introductio

0.1

Naive

Naive

Point-Value Complex Root

NTT

Application

Audio 2D Convolution

Separable Kern

Conclusion

Sample Problems

Past Lecture

Deferences

4 Solution: editorial.

Fair warning, time bounds are ridiculous.

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Root Iterative

Application:

2D Convolutions
Separable Kerne

onclusior

Sample Problems

Past Lecture

References

5 Codeforces Round #296: Given two strings T, S and an error bound k, find all the positions where T occurs in S, where T "occurring" at some index i means that the jth character of T has a corresponding character within k of its position.

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Roots

Iterative

Application

Audio 2D Convolution Separable Kerne

onclusion

Sample Problems

Past

Deferences

5 Solution: Honestly no clue but it has the "FFT" tag.

Examples of problems using the FFT

FFT

Huar

Introduction

Algorithn

Naive

Point-Value Complex Root Iterative

Applications

2D Convolution

onclusion

Sample Problems

Past Lecture

References

6 String matching with wildcards: Given two binary strings T, S, T has length N and has wildcards which match any character in S, find all occurrences of T in S.

Examples of problems using the FFT

FFT

Huar

Introductio

Naive
FTT
Point-Value
Complex Root
Iterative
NTT

Application

2D Convolutions
Separable Kerne

Conclusion

Sample Problems

Past Lecture

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Solution: Encode 1 as 1 and 0 as -1. The dot product between T and the slice that T overlaps with S will be be N if they match exactly and less than N if they don't match exactly. To account for wildcards, encode a wildcard as 0 and count the number of wildcards, C. Then, if they match exactly it will be N - C, and less then that if they don't.

Examples of problems using the FFT

FFT

Huar

Introduction

Algorithm

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

Deferences

This can be generalized to non-binary strings if you apply the above algorithm to each character, setting that character as 1 and not that character as -1. Sum over all possible characters, and that will tell you whether there is a mismatch somewhere (similar to SPOJ MAXMATCH).

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

Deferences

6 This idea can also be applied to string matching without wildcards. Encode each character as its ASCII value in a polynomial, and compute the ℓ^2 -norm between T and S. The ℓ^2 norm will be 0 if they match, and positive if they don't.

Examples of problems using the FFT

FFT

Huar

Introduction

Algorithm

Naive

Point-Value
Complex Roots
Iterative

Application

Audio
2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

Deferences

3 3SUM: Given a list of integers between -N and N, find 3 numbers that add up to 0 (duplicates are allowed).

Examples of problems using the FFT

FFT

Huar

Introductio

0.1

Naive FTT Point-Value Complex Roo

Application

2D Convolution

onclusion

Sample Problems

Past Lecture

References

7 Solution: The basic idea will be to encode the list into a length 2N polynomial p where the degree is an integer value and the coefficient is whether that value appears in the array. Compute p^3 and read off the coefficient of x^0 .

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

Naive

Point-Value Complex Root Iterative

Application

2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

References

Thowever, this doesn't work if the degrees are negative. If the most negative power of x in p is x^{-N} , We can simply multiply p by x^N to make every power positive, making a new polynomial p'. Then, after computing $(p')^3$, instead of looking at the coefficient of x^0 , we can look at the coefficient of x^{-3N} (accounting for the fact that $p' = x^N p$, $(p')^3 = x^{3N} p^3$, $p^3 = \frac{(p')^3}{x^{3N}}$

Examples of problems using the FFT

FFT

Huar

IIItroductio

Algorithms

Naive

Point-Value Complex Roots Iterative

Application

Audio 2D Convolution Separable Kerne

onclusion

Sample Problems

Past

Deferences

7 Alternative solution, if duplicates aren't allowed: here (look for "color coding").

Examples of problems using the FFT

FFT

Huan

Introductio

Algorithm

Naive

Point-Value Complex Root

NTT

Application

Audio 2D Convolution Separable Kerne

onclusion

Sample Problems

Past Lecture

References

Anime Music Quiz: Guess which anime an intro/outro comes from.

Examples of problems using the FFT

FFT

Huar

Introductio

Algorithm

FTT

Point-Value
Complex Root

NTT

Application

Audio
2D Convolutions
Separable Kerne

onclusion

Sample Problems

Past Lecture

Deferences

8 Solution: The Shazam algorithm.

Past Lectures

Past Lectures on similar topics

FFT

Huar

Introductio

Algorithms

Naive

Point-Value Complex Roots Iterative

Applications

2D Convolutions Separable Kerne FFT Algorithm

Conclusion

Sample Problem

Past Lectures

References

- 1 "Edge Detection", (Alexey Didenkov, 2018)
- 2 "Fast Multiplication: Karatsuba and FFT" (Haoyuan Sun, 2016)
- 3 "Multiplying Polynomials", (Haoyuan Sun, 2015)
- 4 "Fast Fourier Transform", (Sreenath Are, 2013)

References

Resources that were useful when compiling this lecture

FFT

Huar

Algorithm:

FTT
Point-Value

Complex Roots
Iterative

Applications

2D Convolutions
Separable Kernels

Conclusion

Sample Problem

Past Lectures

References

- 1 Introduction to Algorithms, chapter 30 (very helpful)
- 2 The number theoretic transform
- μ -law algorithm
- 4 Picard's Existence and Uniqueness Theorem
- **5** Separable convolutions